

Constraints on the genesis and evolution of alkaline and carbonatitic rocks

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M.Sc. Simon Johannes Braunger
aus Biberach an der Riß

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Prof. Dr. József Fortágh

1. Berichterstatter:

Prof. Dr. Gregor Markl

2. Berichterstatter:

PD Dr. Michael A. W. Marks

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Abstract

Alkaline complexes comprise a broad spectrum of mantle-derived silicate rocks which are of considerable scientific and economic interest. They reflect some of the most evolved igneous rocks on Earth, and may contain attractive levels of Rare Earth Elements (REE) and other High Field Strength Elements (HFSE) such as Zr, Hf, Nb, Ta. Some of these complexes are additionally associated with carbonatites. However, many questions regarding the crystallization conditions and processes of such peculiar rock types are still unanswered. This thesis focuses on two mineralogically contrasting alkaline complexes (and nearby localities), namely the Tarosero Volcanic Complex in northern Tanzania, and the Kaiserstuhl Volcanic Complex in Southwest Germany.

The Tarosero volcano is one of the very few complexes that comprises extrusive agpaitic rocks. Early crystallization of OH-bearing amphibole prevented the exsolution of aqueous fluids, thus enabling the enrichment of halogens, REE/HFSE, and the late-stage formation of eudialyte. The variable evolved rocks formed under low redox conditions ($\Delta\text{FMQ} \leq 0$) and derived from a basaltic magma. Moreover, in the nearby Ogo basalt, multistage-magma mixing was demonstrated by various xenocryst phases.

The rocks of the Kaiserstuhl Volcanic Complex formed under high redox conditions ($\Delta\text{FMQ} = +1$ to $+2$) and derived from two parental magmas. Fractional crystallization of a basanitic magma produced the tephritic to phonolitic rock series, while fractional crystallization of an olivine melilititic magma with variable CO_2 concentrations resulted in the formation of the nephelinitic to limburgitic, or the melilititic to h a ynitic rock series and the carbonatites. The magmatic to hydrothermal evolution of the carbonatites can be deciphered by textural and compositional variations in pyrochlore. Furthermore, unusual high amounts of mica in one carbonatite body (Badberg) indicate a steady and continuous supply of elements due to the interaction with the wall rock. The decomposition of silicate xenoliths raised the silica activity of the carbonatite magma, which additionally enabled the crystallization of clinopyroxene, and enhanced the incorporation of REE in apatite. However, despite the high geochemical contrast between primitive dyke magmas and potash salt rocks (at Buggingen), there is no evidence for magmatic interaction processes, presumably due to the relatively dry orthomagmatic melt conditions and the fast cooling of the dyke magma.

In addition, based on a worldwide compilation, combined with new results from seven carbonatite complexes (Kaiserstuhl, Sokli, Kovdor, Palabora, Oka, Magnet Cove, Jacupiranga), it was demonstrated that carbonatites and their associated silicate rocks indicate significantly higher magmatic redox conditions than alkaline rocks without association to carbonatites. This presumably is attributed to the prerequisite of a carbonate-bearing, and hence relatively oxidized mantle source for the genesis of carbonatites.

Zusammenfassung

Alkaligesteinskomplexe umfassen ein breites Spektrum an aus Mantelschmelzen entstandenen Silikatgesteinen, die von beträchtlichem wissenschaftlichen und wirtschaftlichen Interesse sind. Sie stellen einige der am weitesten entwickelten magmatischen Gesteine der Erde dar und können attraktive Gehalte an Metallen der Seltenen Erden (REE) und anderen „High-Field-Strength“-Elementen (HFSE) enthalten (z.B. Zr, Hf, Nb, Ta). Manche dieser Komplexe sind zusätzlich mit Karbonatiten assoziiert. Viele Fragen bezüglich der Kristallisationsbedingungen und -prozesse solch ungewöhnlicher Gesteine sind jedoch noch unbeantwortet. Diese Arbeit konzentriert sich auf zwei mineralogisch gegensätzliche Alkalikomplexe (und nahe gelegene Lokalitäten), den Tarosero-Vulkankomplex in Nordtansania und den Kaiserstuhl-Vulkankomplex in Südwestdeutschland.

Der Vulkan Tarosero ist einer von nur wenigen Komplexen, die extrusive agpaitische Gesteine aufweisen. Die frühe Kristallisation von OH-haltigen Amphibolen verhinderte die Entweichung wässriger Fluide und ermöglichte so die Anreicherung von Halogenen, REE/HFSE, sowie die Bildung von spätmagmatischem Eudialyt. Die unterschiedlich hoch entwickelten Gesteine bildeten sich bei niedrigen Redoxbedingungen ($\Delta FMQ \leq 0$) und stammen von einem basaltischen Magma ab. In den nahegelegenen Ogoi-Basalten konnte zudem anhand verschiedener Fremdmineraleinschlüsse ein mehrstufiger Magmamischungsprozess aufgezeigt werden.

Die Gesteine des Kaiserstuhl-Vulkankomplexes bildeten sich unter hohen Redoxbedingungen ($\Delta FMQ = +1$ bis $+2$) und stammen von zwei verschiedenen Stammmagmen ab. Fraktionierte Kristallisation eines basanitischen Magmas führte zur Bildung einer tephritischen bis phonolithischen Gesteinsserie, während die Fraktionierung eines olivinmelilithischen Magmas mit variablen CO_2 -Konzentrationen zur Bildung der nephelinitischen bis limburgitischen Gesteine, beziehungsweise der melilithischen bis haüynitischen Gesteinsserie inklusive der Karbonatite führte. Die magmatische bis hydrothermale Entwicklung der Karbonatite konnte durch textuelle und mineralchemische Unterschiede im Pyrochlor aufgezeigt werden. Zudem weisen ungewöhnlich hohe Glimmeranteile in einem Karbonatitkörper (Badberg) auf eine stetige und kontinuierliche Elementzufuhr aufgrund der Wechselwirkung mit dem Nebengestein hin. Die Zersetzung von silikatischen Xenolithen erhöhte zudem die

SiO₂-Aktivität des Karbonatitmagmas, was die Kristallisation von Klinopyroxen ermöglichte und einen erhöhten Einbau von Seltenen Erden (REE) in Apatit bewirkte. Trotz des hohen geochemischen Kontrastes zwischen primitiven Ganggesteinsmagmen und Kalisalzgesteinen (bei Buggingen) gibt es jedoch vermutlich aufgrund der relativ trockenen orthomagmatischen Schmelzbedingungen und einer schnellen Abkühlung des Magmas keinen Hinweis für magmatische Interaktionsprozesse.

Darüber hinaus konnte durch das weltweite Zusammentragen von Daten kombiniert mit neuen Ergebnissen aus sieben Karbonatitkomplexen (Kaiserstuhl, Sokli, Kovdor, Palabora, Oka, Magnet Cove, Jacupiranga) gezeigt werden, dass Karbonatite und die mit ihnen assoziierten Silikatgesteine signifikant höhere magmatische Redoxbedingungen anzeigen als Alkaligesteine ohne Assoziation zu Karbonatiten. Dies kann auf eine Karbonat-haltige und damit relativ oxidierte Mantelquelle zurückgeführt werden, die als notwendige Voraussetzung zur Bildung von Karbonatiten angesehen wird.

List of publications in the dissertation

The present thesis deals with the publications listed below. The attached tables provide information about the authors' own contributions to the publications.

Study 1

Zaitsev, A. N., Arzamastsev, A. A., Marks, M. A. W., **Braunger S.**, Wenzel, T., Spratt, J., Salge, T., and Markl, G., submitted. Multiple sources for basaltic volcanism in Laetoli area (Crater Highlands, Tanzania): evidence from mineralogy and geochemistry of the Ogol lavas. *Journal of Petrology*.

Status of the publication process: submitted

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Zaitsev, A. N.	1	30	5	30	30
Arzamastsev, A. A.	2	30	5	30	30
Marks, M. A. W.	3	10	5	10	20
Braunger, S.	4	10	10	10	10
Wenzel, T.	5	10	35	0	0
Spratt, J.	6	0	30	5	0
Salge T.	7	0	5	5	0
Markl, G.	8	10	5	10	10

Study 2

Braunger, S., Marks, M. A. W., Zaitsev, A. N., Wenzel, T., and Markl, G., submitted. The Petrology of the Tarosero Volcanic Complex: Constraints on the formation of extrusive agpaite rocks. *Journal of Petrology*.

Status of the publication process: submitted

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Braunger, S.	1	75	95	70	75
Marks, M. A. W.	2	15	0	15	20
Zaitsev, A. N.	3	0	0	5	0
Wenzel, T.	4	0	5	5	0
Markl, G.	5	10	0	5	5

Study 3

Braunger, S., Marks, M.A.W., Walter, B.F., Neubauer, R., Reich, R., Wenzel, T., Parsapoor, A., Markl, G., 2018. The Petrology of the Kaiserstuhl Volcanic Complex, SW Germany: The Importance of Metasomatized and Oxidized Lithospheric Mantle for Carbonatite Generation. Journal of Petrology 59, 1731-1762. <http://dx.doi.org/10.1093/petrology/egy078>.

Status of the publication process: accepted and published

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Braunger, S.	1	50	70	55	60
Marks, M. A. W.	2	30	0	20	20
Walter, B. F.	3	5	0	5	5
Neubauer, R.	4	0	10	5	0
Reich, R.	5	0	10	5	0
Wenzel, T.	6	0	5	5	5
Parsapoor, A.	7	0	5	0	5
Markl, G.	8	15	0	5	5

Study 4

Walter, B.F., Parsapoor, A., **Braunger, S.**, Marks, M.A.W., Wenzel, T., Martin, M., Markl, G., 2018. Pyrochlore as a monitor for magmatic and hydrothermal processes in carbonatites from the Kaiserstuhl volcanic complex (SW Germany). Chemical Geology 498, 1-16. <http://dx.doi.org/10.1016/j.chemgeo.2018.08.008>.

Status of the publication process: accepted and published

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Walter, B. F.	1	50	55	55	40
Parsapoor, A.	2	0	10	0	0
Braunger, S.	3	10	5	5	10
Marks, M. A. W.	4	30	10	30	30
Wenzel, T.	5	0	20	0	10
Martin, M.	6	0	0	5	0
Markl, G.	7	10	0	5	10

Study 5

Giebel, R.J., Parsapoor, A., Walter, B.F., **Braunger, S.**, Marks, M.A.W., Wenzel, T., Markl, G., 2019. Evidence for Magma–Wall Rock Interaction in Carbonatites from the Kaiserstuhl Volcanic Complex (Southwest Germany). *Journal of Petrology* 60, 1163-1194. <http://dx.doi.org/10.1093/petrology/egz028>.

Status of the publication process: accepted and published

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Giebel, R. J.	1	24	0	34	44
Parsapoor, A.	2	0	50	2	0
Walter, B. F.	3	21	30	15	7
Braunger, S.	4	22	0	24	12
Marks, M. A. W.	5	23	0	20	32
Wenzel, T.	6	0	20	0	0
Markl, G.	7	10	0	5	5

Study 6

Braunger, S., Scharrer, M., Marks, M. A. W., Wenzel, T., Markl, G., submitted. Hydrothermal halite precipitation in mafic volcanic rocks: the case of the Buggingen salt deposit, SW Germany. *Canadian Mineralogist*.

Status of the publication process: submitted

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Braunger, S.	1	50	90	60	75
Scharrer, M.	2	15	0	20	5
Marks, M. A. W	3	10	0	10	10
Wenzel, T.	4	0	10	0	5
Markl, G.	5	25	0	10	5

Study 7

Braunger, S., Marks, M.A.W., Wenzel, T., Chmyz, L., Guitarrari Azzone, R., Markl, G., 2020. Do carbonatites and alkaline rocks reflect variable redox conditions in their upper mantle source? *Earth and Planetary Science Letters* 533, 116041. <http://dx.doi.org/10.1016/j.epsl.2019.116041>.

Status of the publication process: accepted and published

Author	Author position	Scientific ideas (%)	Data generation (%)	Analysis and interpretation (%)	Paper writing (%)
Braunger, S.	1	70	90	75	70
Marks, M. A. W.	2	20	0	15	20
Chmyz, L.	3	0	5	0	0
Guitarrari Azzone, R.	4	0	0	0	5
Wenzel, T.	5	0	5	5	0
Markl, G.	6	10	0	5	5

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Appendix

Submitted manuscripts and accepted publications	Appendix I, II, III, IV, V, VI, VI
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1 Introduction

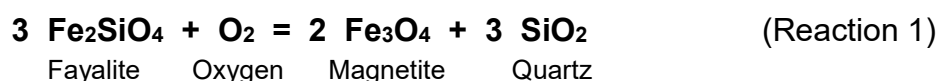
This thesis was prepared at the Department of Earth Sciences at the Eberhard Karls University Tübingen and deals with the crystallization conditions and processes of variably evolved alkaline silicate rocks, and carbonatites. The resulting findings lead to a better understanding of alkaline magmatic systems in general and provide insights into the mantle source regions as well.

1.1 Thermodynamic terms

The thermodynamically stable mineral assemblage of a magmatic system depends on the intrinsic state variables pressure and temperature, as well as on its chemical composition. However, intrinsic parameters like the oxygen fugacity, the silica activity and the water activity are commonly used to describe a magmatic system in more detail and to depict magmatic processes, including e.g. fluid exsolution/degassing, fractional crystallization, melting from variable mantle sources or crustal contamination (e.g. Markl et al., 2001; Halama et al., 2004; Marks et al., 2008).

While fugacities are defined as the thermodynamically effective partial pressure of a system, activities represent the thermodynamically effective concentration of a component in a system. Besides experimentally calibrated approaches (e.g. Gudmundsson and Wood, 1995; Ridolfi and Renzulli, 2012), fugacities and activities can be calculated by appropriate in equilibrium mineral assemblages at a given pressure and temperature. The calculations are based on the minimization of the Gibbs free energy and may be executed by software like Perple_X (Connolly, 1990) or QUILF (Andersen et al., 1993), using internally consistent thermodynamic databases (e.g. Robie and Hemingway, 1995; Holland and Powell, 1998).

Since the absolute oxygen fugacity of a system typically strongly depends on the temperature, redox-sensitive buffer reactions are often used as a reference, such as the reaction (1) between fayalite, magnetite and quartz (FMQ buffer):



The logarithmic deviation of the oxygen fugacity from this buffer (ΔFMQ) is a common measure for the oxidation state of a magmatic system.

1.2 Alkaline silicate rocks: Definition, occurrence and formation

Alkaline silicate rocks comprise a broad spectrum of primitive (e.g. alkali basalt, olivine nephelinite/melilitite, basanite), intermediate (e.g. tephrite, foid monzogabbro) and evolved (e.g. phonolite, foid syenite, rhyolite, granite), mantle-derived rock types. They can be defined as those in which the content of alkali metals compared to silica and/or aluminum is too high to be solely incorporated into alkali feldspar. Thus, they contain foids, sodic pyroxenes, sodic amphiboles or other alkali-rich mineral phases (Fitton and Upton, 1987). Alkaline rocks are mainly present at continental rift systems and in intraplate continental provinces, but have also been reported at ocean-related settings and subduction zone environments (Wooley, 1987; Kogarko et al., 1995; Woolley, 2001, 2019). Although alkaline rock occurrences account for only ~1 % (~3300 localities) of all igneous rocks, they attained much attention over the last decades. This is largely due to their vast mineralogical and compositional variability, with some of them containing attractive levels for REE (Rare Earth Elements) and other HFSE (High Field Strength Elements; e.g. Kogarko, 1990; Goodenough et al., 2016; Smith et al., 2016).

1.2.1 Peralkaline rocks

Peralkaline rocks are defined by a molar $(\text{Na}+\text{K})/\text{Al}$ ratio >1 , and form a volumetrically minor group of alkaline rocks with silica oversaturated to silica undersaturated compositions, such as granites/rhyolites, syenites/trachytes and nepheline syenites/phonolites (e.g. Le Maitre et al., 2002; Frost and Frost, 2008). They are typically characterized by high concentrations of halogens, Large Ion Lithophile Elements (LILE; e.g. Li, Na, K, Rb, Cs), REE and other HFSE (e.g. Zr, Hf, Nb, Ta). Peralkaline rocks can be further subdivided according to their mineralogy: the more common miaskitic rocks contain well-known HFSE-bearing minerals like zircon, baddeleyite or titanite, while agpaitic rocks are defined by rather unusual Na-Ca-HFSE and halogen-bearing minerals such as eudialyte and wöhlerite (Sørensen, 1997; Marks et al., 2011; Marks and Markl, 2017).

The high concentrations of incompatible elements in peralkaline rocks are assumed to derive by partial melting of a metasomatically pre-enriched (sub)lithospheric mantle (e.g. Fitton and Upton, 1987; Martin et al., 1994; Upton et al., 2003; Kogarko et al., 2010; Markl et al., 2010), with subsequent extensive fractional crystallization of the

parental basaltic, basanitic or nephelinitic magma (e.g. Marks and Markl, 2017 and references therein). Agpaitic rocks represent the most evolved stage and are characterized by the highest abundances of REE/HFSE in peralkaline systems. Volatile elements like alkali metals and halogens increase the solubility of REE and other HFSE (Watson, 1979; Keppler, 1993; Linnen and Keppler, 2002) and thus, need to be retained in the melt in order to form agpaitic minerals. This is generally reached by a combination of the following processes:

- a low water activity which inhibits the expulsion of alkali- and halogen-rich fluids (e.g. Kogarko, 1987; Markl et al., 2001; Marks and Markl, 2015)
- magma evolution under low redox conditions. This suppresses the early crystallization of magnetite, thus leading to an increase of the Fe concentration, which in turn enhances the solubility of halogens (Metrich and Rutherford, 1992; Giehl et al., 2013). Moreover, low redox conditions may prevent the escape of aqueous fluids by stabilizing CH_4+O_2 instead of $\text{CO}_2+\text{H}_2\text{O}$ mixtures (Markl et al., 2001)
- crystallization under closed system conditions, such as plutons, dykes and domes (Sørensen, 1992)

The last mentioned process is in accordance with the fact that almost all of the approximately 100 agpaitic occurrences represent intrusive rocks (Marks and Markl, 2017). Very rare extrusive peralkaline rocks with late stage agpaitic minerals (e.g. at Tarosero, Tanzania; Dawson, 1997) demonstrate a formation under open system conditions, and therefore require peculiar crystallization conditions. Moreover, agpaitic complexes mostly lack carbonatites, thus indicating contrasting formation processes compared to carbonatite complexes.

1.3 Carbonatites: Definition, occurrence and formation

Carbonatites are rocks of magmatic origin with a modal primary carbonate content of at least 30 vol.-% (Mitchell, 2005). Most occurrences (~60 %; Chakhmouradian et al., 2016) are mineralogically classified as calcite-carbonatites, and can be divided into coarse-grained carbonatites (sövites), and medium to fine-grained carbonatites (alvikites; Le Maitre et al., 2002). Other carbonatite types include dolomite-carbonatites (beforsites), ferrocarnatites and natrocarbonatites, with the latter occurring at the only active carbonatite complex, the Oldoinyo Lengai in Tanzania (Dawson, 1962;

Keller and Krafft, 1990). Besides carbonate, carbonatites often consist of many other mineral phases, including phosphates (predominantly apatite), silicate minerals (e.g. olivine, clinopyroxene, mica, garnet), oxides (magnetite, ilmenite, pyrochlore, perovskite, baddeleyite) and sulfides (pyrrhotite, galena, pyrite). Some of these minerals might have formed due to the assimilation of wall rock, or they represent entrained xenocrysts (Barker, 2001; Vuorinen and Skelton, 2004; Chakhmouradian et al., 2008). Because of the presence of apatite and REE/HFSE-rich minerals like pyrochlore, perovskite and baddeleyite, carbonatites are not only of scientific interest but also provide economic potential (e.g. Kanazawa and Kamitani, 2006; Smith et al., 2016).

To date, there are about 550 known carbonatite occurrences, with over 90 % being intrusive rocks (Woolley and Kjarsgaard, 2008). About 80 % of the carbonatites are spatially and temporally associated with alkaline silicate rocks, suggesting a close genetic relationship between both rock types. Most of these silicate rocks are SiO₂-undersaturated and comprise variable primitive to evolved compositions. Mitchell (2005) divides carbonatites according to the dominant silicate rocks of the complex into common (I) nephelinite and (II) melilitite clan carbonatites, and rare (III) kimberlite, (IV) aillikite and (V) peralkaline (agpaitic) nephelinite clan carbonatites. Note that subvolcanic and plutonic equivalents are also included in this classification.

It is generally accepted that carbonatites and most of their associated alkaline silicate rocks have their origin in the lithospheric or sublithospheric (asthenospheric) mantle (Bell and Simonetti, 2010). Two hypotheses are discussed for the formation of carbonatites. They either form by low degree partial melting of a carbonated peridotite, leading directly to carbonatitic melts (Wallace and Green, 1988; Harmer et al., 1998; Ying et al., 2004), or by fractional crystallization of a carbonate-bearing, typically melilititic or nephelinitic, silicate melt. The latter process results in the genesis of carbonatites by the melt reaching its silicate-carbonate liquidus boundary (Lee and Wyllie, 1994, 1998), or by exceeding the carbonate solubility with consequent exsolution of a carbonatitic melt (Kjarsgaard and Peterson, 1991; Halama et al., 2005; Brooker and Kjarsgaard, 2011). These primary magmatic carbonatites are distinguished from so-called carbothermal carbonatites which form by CO₂-dominated residual fluids originating from a wide range of magmas (Mitchell, 2005). In some

cases, carbonatites are associated with phoscorites, ultramafic apatite-magnetite rocks which additionally contain forsterite, diopside or phlogopite, and commonly also carbonate (Russell, 1954; Krasnova et al., 2004b and references therein). Phoscorites are genetically related to carbonatites by crystal fractionation or liquid immiscibility from a common phosphate- and carbonate-rich magma (Rimskaya-Korsakova and Krasnova, 2002; Krasnova et al., 2004a; Lee et al., 2004; Giebel et al., 2019).

1.4 Redox conditions of the mantle

As stated above, alkaline magmatism is related to partial melting of the (sub)lithospheric mantle. Besides pressure and temperature, the melting point of the mantle strongly depends on phases such as H₂O, CO₂ and carbonate (Dasgupta and Hirschmann, 2006; Foley et al., 2009). However, whether these rather “oxidized” phases, or “reduced” phases such as graphite/diamond (C), carbides or CH₄ are stable in the mantle, is mainly controlled by the oxygen fugacity (Stagno, 2019). The redox conditions of the lithospheric mantle vary from $\Delta\text{FMQ}=-4$ to $\Delta\text{FMQ}=+4$, with considerable differences between various geodynamic settings (Frost and McCammon, 2008; Foley, 2011). Lithospheric mantle xenoliths found within cratons indicate relatively low redox conditions ($\Delta\text{FMQ}=-2.8$) which are related to the pressure-dependent stability of reduced Fe²⁺-bearing and oxidized Fe³⁺-bearing mineral phases. The redox state of the mantle thus decreases by approximately 0.7 ΔFMQ units per Gigapascal (GPa) and is particularly low in the deeper parts of thick cratons (Ballhaus and Frost, 1994). Mantle xenoliths found at subduction zones are on average significantly more oxidized ($\Delta\text{FMQ}=+0.5$) due to the interaction with slab-derived melts. The oxidative character of these melts may be caused by the oxidized subducted material (Foley, 2011), or by the fact that during partial melting, trivalent iron preferably partitions into the melt while divalent iron remains in its residue (Canil and O'Neill, 1996; Mallmann and O'Neill, 2009).

Although no asthenospheric xenoliths are recorded, it is assumed that the asthenosphere is slightly more reduced than the lithosphere because of the loss of oxidized melts and fluids which interact with the overlying mantle (Frost and McCammon, 2008). This is consistent with redox estimations on abyssal peridotites and redox calculations on mid-ocean ridge basalts, which are used as an approximation for the oxidation state of the asthenosphere (Christie et al., 1986; Bézou

and Humler, 2005; Foley, 2011). A relatively high redox state of the asthenospheric mantle is indicated by the redox conditions of ocean island basalts and associated xenoliths ($\Delta\text{FMQ}=+0.6$; Ballhaus, 1993; Mallmann and O'Neill, 2007; Foley, 2011; Eguchi and Dasgupta, 2018). This might be related to the influence of subducted (oxidized) lithologies on the genesis of ocean island basalts (e.g. Weaver, 1991).

1.5 Redox conditions of alkaline silicate rocks and carbonatites

Redox estimates on magmatic rocks are assumed to provide deep insights into the oxidation state of the source region (Carmichael, 1991; Cottrell et al., 2020), although factors such as the melting degree, degassing processes or fractional crystallization complicate a direct link to the mantle source (Gaillard et al., 2015). Similar to the (sub)lithospheric mantle, the redox conditions during the formation of alkaline silicate rocks are highly diverse and vary between $\Delta\text{FMQ}=-4$ and $\Delta\text{FMQ}=+5$ (Carmichael, 1991; Marks and Markl, 2017). This redox variation is qualitatively visible in the mineralogy of the rocks, with clinopyroxene-garnet-titanite assemblages typically indicating high redox conditions above the FMQ buffer. Alkaline rocks with olivine, clinopyroxene and Fe-Ti oxides can demonstrate both oxidized and reduced formation conditions, while agpaitic rocks with their exotic minerals such as eudialyte or wöhlerite belong to the most reduced alkaline rocks (Marks and Markl, 2017).

Little is known about the redox conditions of carbonatites and their associated silicate rocks. The few previous studies either concentrate solely on the carbonatites (e.g. Mud Tank, Australia; Currie et al., 1992) or the silicate rocks (e.g. Tamazeght, Morocco; Marks et al., 2008). Many of these studies also deal with late to post-magmatic processes (Milani et al., 2016; Ivanyuk et al., 2017), or focus on rather atypical carbonatite occurrences, such as the graphite-bearing carbonatites of Newania (India; Doroshkevich et al., 2010) and Chagatai (Uzbekistan; Kogarko and Ryabchikov, 2013).

2 Objective

Alkaline silicate rocks and carbonatites are of great scientific interest due to the large spectrum of mineralogically variable rock types, and attained economic attention as an exploration target for “hi-tech” raw materials like REE and other HFSE (e.g. Zr, Hf, Nb, Ta). However, especially due to the large range of different rock types and their rarity with respect to igneous rocks in general, there are many unanswered questions concerning their crystallization conditions and processes.

Ideal places to investigate the formation and magmatic evolution of alkaline rocks are the Crater Highlands and the adjacent Gregory Rift in northern Tanzania. This region is known for its intense alkaline magmatism, with volcanoes typically belonging to the nephelinite/melilitite \pm carbonatite series, or the basalt-trachyte series.

The Ogol lavas are considered as the most primitive basaltic rocks in the Crater Highlands, but despite their importance for the understanding of alkaline magmatism in this region, they are only poorly investigated. Study 1 (Chapter 4.1) provides a detailed petrological investigation of the Ogol lavas and focuses on:

- evaluating the role of crustal contamination and magma mixing;
- the genetic relationship between the Ogol lavas and the nearby volcanoes;
- the crystallization conditions (pressure, temperature).

Agpaitic rocks generally demonstrate a formation under closed system conditions (e.g. plutons, dykes, domes). However, the Tarosero Volcanic Complex (Gregory Rift, Tanzania) is one of the very few localities where extrusive agpaitic rocks have been reported. In order to shed light on the prerequisites to form such unusual rock types, Study 2 (Chapter 4.2) focuses on:

- the crystallization conditions of the different rock types at Tarosero (pressure, temperature, oxygen fugacity, silica activity, water activity);
- the genetic relationship between the various rock types.

The Kaiserstuhl Volcanic Complex (Germany) is a well-suited locality to study the evolutionary history of a carbonatite complex since it consists of various types of alkaline silicate rocks (e.g. nephelinites, tephrites, phonolites) and plutonic to volcanic facies of carbonatites. However, despite the broad spectrum of different lithologies,

there is no detailed petrological study for this volcano. The first study on the Kaiserstuhl (Study 3; Chapter 4.3) therefore deals with:

- the determination of the formation conditions of carbonatites and silicate rocks (pressure, temperature, oxygen fugacity, silica activity);
- the genetic relationship between the various rock types;
- the constraints of the mantle source(s).

The large crystallization interval of ubiquitous minerals like pyrochlore provides potential for a very detailed insight into the evolution of carbonatite magmas. The second study on the Kaiserstuhl (Study 4; Chapter 4.4) focuses on:

- the textural and geochemical differentiation of the pyrochlore types;
- the processes leading to the different pyrochlore types.

A high geochemical contrast between magmas and wall rocks allows for intense assimilation processes. However, the interaction between carbonatite magmas and silicate wall rocks is only poorly investigated (Study 5; Chapter 4.5). Likewise, little is known about the influence of salt-rich host rocks on the mineralization of primitive silicate dyke rocks (Study 6; Chapter 4.6). The studies on the Kaiserstuhl (Study 5) and at the nearby Buggingen (Study 6) deal with:

- the mineralogical/mineral chemical evidence for contamination/assimilation;
- the discrimination between magmatic and hydrothermal interaction processes;
- the influence of contamination/assimilation on the crystallizing mineral phases;
- the characterization of the mineral precipitation mechanism.

Little is known about the redox conditions during formation of carbonatites and their spatially associated alkaline silicate rocks. However, redox conditions of igneous rocks may provide insight into their mantle source and may help to understand the peculiar prerequisites to form carbonatites. Therefore, 7 carbonatite complexes (Kaiserstuhl, Sokli, Kovdor, Palabora, Oka, Magnet Cove, Jacupiranga) were investigated in Study 7 (Chapter 4.7), which concentrates on:

- the determination of the redox conditions for carbonatites and silicate rocks;
- the distinction between magmatic and post-magmatic redox conditions;
- the comparison of the redox conditions between alkaline complexes with and without associated carbonatites;
- the influence of the mantle redox state on the formation of alkaline rocks.

3 Selected study areas

The selected study areas for this thesis concentrate on two regions of alkaline magmatism: i) the North Tanzanian Divergence Zone where the Ogot lavas in the Laetoli area (Study 1) and the Tarosero Volcanic Complex (Study 2) were investigated, and ii) the Upper Rhine Graben in Southwest Germany with focus on the Kaiserstuhl Volcanic Complex (Studies 3-5) and the primitive dyke rocks at Buggingen (Study 6). Moreover, besides the Kaiserstuhl Volcanic Complex, six further carbonatite complexes were investigated (Study 7). These include Sokli (Finland) and Kovdor (Russia) on the Kola Peninsula, Palabora in South Africa, Oka in Canada, Magnet Cove in the USA, and Jacupiranga in Brazil. The differences in the formation age (Palaeoproterozoic to Neogene), the emplacement depth (plutonic, subvolcanic and volcanic), and the mineralogy of the carbonatites and associated silicate rocks reflect the typical spectrum of carbonatite occurrences worldwide (Mitchell, 2005; Woolley and Kjarsgaard, 2008). A more detailed description of the geology for these 6 complexes is given in Appendix VII.

3.1 Alkaline magmatism in the North Tanzanian Divergence Zone

The eastern branch of the East African Rift System is known for its intense magmatism. Rifting was initiated about 30 Ma ago by a mantle plume beneath Ethiopia, and evolved southwards along Pan-African suture zones (Hofmann et al., 1997; Chorowicz, 2005) towards the North Tanzanian Divergence Zone (Smith, 1994; Le Gall et al., 2008; Mana et al., 2015). Magmatic activity in this zone can be divided into two groups based on major faulting events (Dawson, 1992):

- i) volcanoes older than 1.2 Ma which mainly consist of alkali basalts and trachytes to phonolites (e.g. Tarosero, Monduli, Ngorongoro, Lemagarut, Oldoinyo Sambu, Ketumbeine, Kilimanjaro), and rarely of nephelinites ± carbonatites (e.g. Burko, Sadiman, Essimingor, Mosonik);
- ii) volcanoes younger 1.2 Ma which typically comprise nephelinites to phonolites ± carbonatites (e.g. Meru, Oldoinyo Lengai, Kerimasi, Kwahera, Hanang).

3.1.1 Ogol basalts

The Crater Highlands are situated next to the western escarpment of the Gregory Rift and consist of numerous volcanic centers (e.g. Lemagarut, Ngorongoro, Sadiman) in a ~100 km long zone between Lake Natron and Lake Eyasi. They mark the southern termination of the eastern branch of the East African Rift valley. The oldest magmatic activity in this region, beginning for 4.63 ± 0.05 Ma (Mollet et al., 2011), is attributed to the Sadiman volcano, west of which the Laetoli tuffs with the 3.66 Ma old *Australopithecus afarensis* footprints were discovered (Leakey and Hay, 1979). The ~2.3 Ma old Ogol basalt lavas (Mollet, 2007) erupted through fissures and cinder cones of up to 1.5 km in diameter and 150 m in height (Hay, 1987).

3.1.2 Tarosero Volcanic Complex

The Tarosero volcano is situated in the Gregory Rift, about 40 km northwest of Arusha. It was formed ~2.3 Ma ago (Macintyre et al., 1974; Mana et al., 2015) and extruded on a northerly dipping plateau of alkali basalt and trachyandesite lavas. The lava cone itself mainly consists of trachyte lavas (and rare tuffs), while smaller occurrences of extrusive rocks ranging from primitive basalts over peralkaline trachytes to agpaitic, eudialyte-bearing phonolites have also been described (Dawson, 2008).

3.2 Alkaline magmatism in the Upper Rhine Graben

Alkaline magmatism in the Upper Rhine Graben belongs to the European Cenozoic Volcanic Province (Wilson and Downes, 1991). The opening of the Upper Rhine Graben results from the stress regime induced by the Alpine orogeny and caused lithospheric thinning (Edel et al., 1975; Ziegler, 1982; Dèzes et al., 2004). This, in turn, enabled regional asthenospheric upwelling, and finally led to the emplacement of alkaline rocks along deep-reaching zones of weakness (Hüttner, 1996; Edel et al., 2006; Bourgeois et al., 2007).

3.2.1 Kaiserstuhl Volcanic Complex

The Kaiserstuhl Volcanic Complex is located about 20 km northwest of Freiburg. It was active 19 to 16 Ma ago (Kraml et al., 2006 and references therein) and was fed by several magma sources (Keller, 1978b, 1984a; Keller et al., 1990; Schleicher et al., 1990; Schleicher et al., 1991). A potassium-rich magma source led to the formation of the sodalite monzogabbros in the centre of the complex, and the tephritic rocks which

comprise most parts of the current exposure level. This rock series also includes the phonolite stocks at the central, the western and the eastern part of the volcano. Nosean syenites were rarely found in the central part, and have also been proved by scientific drillings at depth (Kirchheimer, 1973). The second magma source formed the volumetrically minor group of sodium-rich rocks such as the olivine nephelinites and limburgites, which mostly occur at the Limberg-Lützelberg complex near Sasbach. Melilite-bearing dyke rocks and the carbonatites are also believed to be genetically related to the second magma series, and predominantly are present in the central parts of the Kaiserstuhl complex. Besides two smaller occurrences at Kirchberg and Katharinenberg, there is a total of four large sövite bodies at Badberg, Haselschacher Buck, Orberg and Degenmatt. The westward dipping Tuniberg fault produces a vertical offset of ~1000 to ~3000 m, separating the carbonatite at the Badberg from the other three bodies (Groschopf et al., 1996; Beccaletto et al., 2010; Brauch et al., 2018). The banded Badberg carbonatite is inclined by 60° to the northwest, and contains cm- to m-sized silicate xenoliths (Blust, 1993). Extrusive carbonatites occur in the west at Kirchberg and Henkenberg as alternating sequences of carbonatitic tuffs, lavas, lapilli stones and silicate pyroclastics (Keller, 1978a, 1981, 1989). Beforsite and alvikite dykes mainly crosscut the subvolcanic rocks of the centre, but also occur in the distal parts of the complex (Katz and Keller, 1981; Katz-Lehnert, 1989).

3.2.2 Buggingen

Nephelinitic and melilititic dyke rocks frequently occur in the vicinity of the Kaiserstuhl Volcanic Complex, such as the olivine melilitite stock of Mahlberg in the north, or the olivine melilitites and monchiquites at Buggingen in the south (Wimmenauer, 1952; Dunworth and Wilson, 1998). At the latter, the primitive dyke rocks intruded into a potash salt deposit of Rupelian age (34 to 28 Ma; Braitsch et al., 1964; Sissingh, 1998). The deposit was mined at a depth level of 800 to 1000 m and belongs to the Upper Saliferous Zone, consisting of sylvite, halite, anhydrite, and marl layers (Esslinger, 1976). It is underlain by calcite- and dolomite-bearing marls (Middle Saliferous Zone), and overlain by anhydrite-, gypsum-, and halite-rich marls. The uppermost part of this sequence consists of bituminous marine shales (Fish Shale), and marine-brackish to limnic marls, which are cut off by unconformable Quaternary sands and gravel.

4 Individual studies

4.1 Evidence for multi-stage magma mixing in basaltic rocks

Title of publication:

Multiple sources for basaltic volcanism in Laetoli area (Crater Highlands, Tanzania): evidence from mineralogy and geochemistry of the Ogol lavas. (Study 1)

In the Crater Highlands, numerous eruptive centers are spatially and chronologically close together, with lava flows and tuffs of variable and partly unknown origin being intercalated (e.g. Mollel, 2007). Likewise, the intense alkaline magmatism provides potential for complex magma mixing processes at depth. We demonstrate such interaction processes by the Ogol basalts, which were investigated in terms of mineralogy, mineral chemistry (using the electron microprobe, EPMA) and whole-rock geochemistry (X-ray fluorescence analysis, XRF; inductively coupled plasma mass spectrometry, ICP-MS) including isotope analyses (thermal ionization mass spectrometer, TIMS).

The relative depletion of Cs, Rb and K in whole-rock samples (compared to Ocean Island Basalts) indicate a formation of the primary Ogol melts by partial melting of a mantle with residual amphibole and/or phlogopite. Moreover, the $(Tb/Yb)_N$ ratio demonstrates melting in the presence of residual garnet (Wang et al., 2002; Mana et al., 2015). Variable $(La/Sm)_N$ and Sr and Nd initial isotopic ratios in the basalts can be explained by assimilation and fractional crystallization (AFC) between the primary Ogol melts, the Tanzanian Craton granite-gneisses and the involvement of up to 5 % alkaline component. However, no mineralogical evidence for crustal contamination was found in the studied rock samples. Both major and trace element data, as well as Sr-Nd isotope ratios demonstrate that mixing with basaltic to trachyandesitic melts of the spatially associated Lemagarut volcanic system is the more likely process. Such magma mixing processes are in accordance with several xenocryst mineral phases in the Ogol lavas, including e.g. Fe-rich olivine cores, chemically distinct (and sometimes spongy) clinopyroxene varieties and embayed plagioclase. Further xenocryst phases comprise perovskite with ilmenite reaction rims, and schorlomite. Since these minerals are rather unusual in basaltic to trachyandesitic magmas, the involvement of an

alkaline component, possibly related to the nearby Sadiman volcano or the Laetoli tuffs (e.g. Zaitsev et al., 2012), is required. This conforms with highly alkaline, phonolitic glass ($[\text{Na}+\text{K}]/\text{Al}=0.98-1.07$) in the investigated Ogol basalts. The compositionally variable mantles and rims around the clinopyroxene xenocrysts suggest that magma mixing was a multistage process that happened at variable depths. The orthomagmatic crystallization conditions of the clinopyroxenes were determined by applying the clinopyroxene-liquid thermobarometer of Neave and Putirka (2017) using whole-rock compositions as a proxy for the melt composition. Although the basalts contain plenty of xenocrysts, rough estimates might be possible as the required liquid components of the pressure equation (Al, Na, K, Si) may only be slightly influenced by the dominant xenocryst phases olivine and clinopyroxene. The calculations indicate an orthomagmatic clinopyroxene crystallization temperature of $\sim 1150-1220$ °C at an upper-crustal depth of 3-12 km (0.1-0.4 GPa). Overall, the most primitive Ogol sample contains relatively few xenocrystals and might reflect the parental melt for the rocks of the Lemagarut volcano. These rocks are estimated to have evolved by fractionation of 50-60 % primary Ogol melt with addition of 5 % crustal material. However, a more detailed study of the Lemagarut volcano is needed to verify this assumption.

4.2 Prerequisites for the formation of extrusive agpaitic rocks

Title of publication: *The Petrology of the Tarosero Volcanic Complex: Constraints on the formation of extrusive agpaitic rocks.* (Study 2)

This study focuses on the genesis of the Tarosero Volcanic Complex in northern Tanzania, which is one of the very rare examples involving extrusive agpaitic rocks. It mainly consists of trachyte lavas and some trachytic tuffs, but additionally comprises minor occurrences of extrusive basalts, andesites, latites, as well as peralkaline trachytes, olivine trachytes and phonolites, with the latter occasionally containing interstitial eudialyte (Dawson, 2008). The magmatic evolution of this volcano is demonstrated by a mineralogical, mineral chemical (EPMA) and geochemical investigation (XRF; ICP-MS; Combustion Ion Chromatography, CIC) of the different rock types, combined with calculations of the intrinsic parameters and fractional crystallization models.

Intrinsic parameters including pressure, temperature, silica activity and oxygen fugacity were calculated by several thermo(oxy)barometers and in-equilibrium mineral assemblages. Pressure estimates for the different rock types illustrate a multi-level magmatic system. While clinopyroxene phenocrysts in the basalts demonstrate a pressure of 1.1 ± 0.2 GPa (~25-40 km depth) which might reflect magmatic underplating at the Moho (Prodehl et al., 1997), amphiboles of the intermediate andesite and the trachytes demonstrate pressures of 0.7 GPa and 0.9 ± 0.1 GPa, respectively (depths of 20-30 km). Peralkaline rocks typically evolve in a shallow-level plumbing system (~2-4 km depth; Mahood, 1984; Novak and Mahood, 1986; Spandler and Morris, 2016; Marks and Markl, 2017) which is in agreement with coarse-grained cumulates that indicate a nearby magma chamber. The liquidus temperature was determined to be 1000-1290 °C in the basalts and decreases to ~900-1100 °C in the intermediate andesites, latites and trachytes. The groundmass assemblage illustrates temperatures around 850 °C. Experimental constraints for the stability of aenigmatite and arfvedsonite determine the groundmass formation of the peralkaline rocks to 750 ± 50 °C (Ernst, 1962; Bailey, 1969; Thompson and Chisholm, 1969; Scaillet and Macdonald, 2001; Giehl et al., 2013; Giehl et al., 2014), which conforms with the feldspar solvus of ≥ 700 °C. Silica activity calculations mostly yield values of $a_{\text{SiO}_2} = 0.4 \pm 0.2$. The redox conditions typically vary between $\Delta\text{FMQ} = 0$ and $\Delta\text{FMQ} = -1$, while there is a large uncertainty in the peralkaline rocks with values ranging from $\Delta\text{FMQ} = -5$ to $\Delta\text{FMQ} = 0$.

The relatively low K, Cs and Rb abundances in primitive mantle-normalized incompatible element pattern of the primitive and intermediate rock types indicate a derivation of a mantle source with residual amphibole±phlogopite (e.g. Halliday et al., 1995; Mana et al., 2015). A genetic link between the various investigated rocks at Tarosero is suggested by the continuous trend in the mineral composition and the very similar ages for the primitive basalts and the peralkaline rock types (Macintyre et al., 1974; Mana et al., 2015). Moreover, the negative Eu and Sr anomalies of the evolved rock types, as well as the low Sr concentration in eudialyte demonstrate extensive plagioclase fractionation which is typical for a basaltic rather than a basanitic or a nephelinitic parental magma (Marks et al., 2011). Fractional crystallization based on the phenocryst phases (olivine + clinopyroxene) cannot explain the evolution from the basalts to the intermediate latites (nor does the additional fractionation of the

groundmass minerals titanomagnetite + plagioclase). Hence, the genetic relationship between these two rock types remains enigmatic. However, fractional crystallization of variable amounts of olivine, clinopyroxene, plagioclase, magnetite, ilmenite and apatite, accompanied by minor crustal contamination, enables the magmatic evolution from the latites via the trachytes towards the peralkaline trachytes and phonolites. Contamination and assimilation in the evolved rock types is in accordance with trace element modelling and Sr-Nd-Pb isotope data of Mana (2013) and Mana et al. (2015). The agpaitic composition of some phonolites is reached by further fractional crystallization of alkali feldspar and minor amphibole. In contrast to the trachytes and all other peralkaline rocks, the peralkaline olivine trachytes are characterized by olivine phenocrysts but lack clinopyroxene and calcic amphibole phenocrysts. The mineral chemistry of olivine, Fe-Ti-oxides and interstitial amphibole implies a close genetic relation with the intermediate latites. The derivation of the peralkaline olivine trachytes from the latites can also be modelled by fractional crystallization (olivine, clinopyroxene, magnetite, ilmenite, plagioclase) and minor crustal contamination.

Alkali metals and halogens in the melt increase the solubility for REE and other HFSE and support the formation of agpaitic minerals (Watson, 1979; Keppler, 1993; Linnen and Keppler, 2002). In case of the extrusive agpaitic rocks at Tarosero, the early crystallization of OH-bearing amphibole resulted in relatively dry late-magmatic conditions which prevented the exsolution of fluid phases and the consequent loss of water-soluble alkali metals and halogens (e.g. Kogarko, 1987; Markl et al., 2001; Liebscher and Heinrich, 2007; Marks and Markl, 2015). Moreover, the low redox conditions (below FMQ) suppressed intense crystallization of magnetite, hence increasing the Fe concentration of the melt, which additionally enhanced the solubility of Cl (Metrich and Rutherford, 1992; Giehl et al., 2013). Finally, the concentration of REE/HFSE could increase during fractionation, leading to the late-stage crystallization of eudialyte in the most evolved peralkaline phonolites.

In contrast to the peralkaline trachytes/phonolites, the peralkaline olivine trachytes are characterized by much lower REE/HFSE concentrations and the lack of agpaitic minerals. A high water activity is demonstrated by the relatively high OH-contents in the interstitial amphiboles and the absence of clinopyroxene and sodalite (Bailey, 1969; Scaillet and Macdonald, 2001; Andersen and Sørensen, 2005; Giehl et al., 2014). Fractional crystallization under rather low crustal pressures presumably

impeded the transition from olivine to OH-bearing amphibole as the dominant phenocryst phase, thus resulting in an increasing water activity during magma evolution. We suggest that the evolution towards the peralkaline olivine trachytes was accompanied by significant amounts of fluid exsolutions, which prevented the enrichment of Cl, REE and other HFSE, and the final crystallization of eudialyte.

4.3 The evolution of silicate magmas in carbonatite complexes

Title of publication: *The Petrology of the Kaiserstuhl Volcanic Complex, SW Germany: The Importance of Metasomatized and Oxidized Lithospheric Mantle for Carbonatite Generation.* (Study 3)

In this study, we demonstrate the evolutionary history for the Kaiserstuhl Volcanic Complex and give implications for the required conditions to form carbonatite complexes in general. The whole-rock major element composition was determined by XRF, the trace elements by ICP-MS, and the mineral chemistry was analyzed using the electron microprobe (EPMA).

The following four rock groups were distinguished at the Kaiserstuhl based on their mineralogy and geochemistry: (I) primitive nephelinitic to limburgitic rocks; (II) melilititic to haüynitic dyke rocks; (III) tephritic to phonolitic rocks including the monchiquite dykes; (IV) and carbonatites, from which the olivine- and monticellite-bearing varieties were investigated in more detail.

Rock groups I and II are characterized by a molar Na/K ratio > 2 , while the tephritic to phonolitic rocks (group III) typically have a Na/K ratio < 2 . Strontium isotope studies reveal the same division, with the carbonatites being assigned to the Na-rich rock series (Schleicher et al., 1990; Schleicher et al., 1991).

Intrinsic parameters including pressure, temperature, silica activity and oxygen fugacity were determined by various thermo(oxy)barometers and appropriate in-equilibrium mineral assemblages. A subvolcanic emplacement depth of approximately 1 km (~ 0.03 GPa) was calculated for the tephritic rocks based on their amphibole composition. Core analyses typically indicate higher pressure estimates than rim analyses and may reflect a continuous crystallization over a distance of 2–4 km during

the ascent from the magma chamber. The orthomagmatic crystallization temperature decreases from more than 1100 °C in the primitive rocks (olivine nephelinites, monchiquites) towards ~880 °C in the evolved phonolitic rocks and the carbonatites. The silica activity increases from the nephelinitic to the limburgitic rocks ($a_{\text{SiO}_2}=0.4-0.9$), which is in accordance with the absence of plagioclase in the former. The presence of perovskite in the melilititic to haüynitic group demonstrates a silica activity ≤ 0.2 , while a decline of the silica activity is observed from the tephritic to the more evolved phonolitic rocks ($a_{\text{SiO}_2}=0.7-0.2$). In the carbonatites, the absence of clinopyroxene (diopside), but the presence of monticellite indicates a silica activity ≤ 0.05 . Calculated redox conditions for the different rock types typically show high mean values ($\Delta\text{FMQ}=+1$ to $+2$), although lower redox estimates are possible for rocks with an insufficiently well determined lower silica activity constraint.

The importance of fractional crystallization for producing the various lithologies at the Kaiserstuhl was demonstrated by mass balance calculations which are based on the assumption that phenocrysts mainly control the fractionation process. Due to the high formation temperature and the primitive composition, the monchiquite was used as the parental magma for the evolution towards the tephritic and phonolitic rocks. Fractional crystallization of garnet caused a depletion of heavy REE in the remaining melt. In the most evolved phonolitic rocks, fractionation of titanite is mirrored by relatively low concentrations of middle REE (e.g. Nicholls and Harris, 1980; Prowatke and Klemme, 2005). Increasing Zr/Hf, Y/Ho and Nb/Ta ratios during melt evolution strengthen the assumption of garnet and/or titanite fractionation. However, the chemical composition of the most evolved phonolitic rocks requires the additional accumulation of nosean and alkali feldspar.

The olivine melilitite from the Mahlberg north of the Kaiserstuhl is assumed to reflect the parental magma for the olivine nephelinites (Wilson et al., 1995; Dunworth and Wilson, 1998; Ulianov et al., 2007) and hence, was used as the starting composition for the further magma evolution. Depending on the olivine/melilitite fractionation ratio, the magma evolved towards the nephelinitic-limbürgitic (low olivine/melilitite ratio) or the melilititic-haüynitic rock series (high olivine/melilitite ratio). A high CO_2 partial pressure favors the crystallization of melilitite, indicating that the magma evolution is controlled by polybaric melting, or by differences in the redox conditions of the parental magmas

(Mysen et al., 1976; Dunworth and Wilson, 1998). Evolution towards the haüynitic composition additionally requires accumulation of haüyne, clinopyroxene and melilitite. Based on the $^{87}\text{Sr}/^{86}\text{Sr}$ isotope ratio (Schleicher et al., 1990), the carbonatites are assigned to the sodium-dominated rock series, and thus presumably originate from the same magma source. This conforms with the relatively high carbonate content of the haüyne melilitites, which are considered to represent the genetic link between the silicate rocks and the carbonatites (Keller, 1984a; Hubberten et al., 1988). However, liquid immiscibility between the carbonatites and haüyne melilitites illustrating the conjugated silicate melt was ruled out as the hypothetically unexsolved precursor melt (haüyne melilitite+carbonatite) cannot be modelled by fractional crystallization of the parental olivine melilitites. The lack of an appropriate silicate conjugate might be related to decarbonation processes in the silicate magmas. We therefore suppose that the precursor melt for the carbonatites is present at greater depths where the pressure is high enough to suppress CO_2 degassing.

The relatively high potassium concentrations for the tephritic to phonolitic rocks, as well as the presence of rare mica xenocrysts (presumably of lithospheric mantle origin) indicate the derivation of a phlogopite- and amphibole-bearing lithospheric mantle (Wilson and Downes, 1991; Wilson and Downes, 1992). Based on the isotope systematics and the trace element chemistry of the olivine melilitites and olivine nephelinites (Wilson et al., 1995; Dunworth and Wilson, 1998; Ulianov et al., 2007), the source for the sodium-rich rock group is assumed to be the lithospheric mantle that had been previously metasomatized by carbonate-rich melts. Since the redox conditions steadily decrease with depth (Ballhaus and Frost, 1994), such carbonate-bearing liquids may be the result of lithospheric thinning: Diamond or graphite that has been stored at the base of the lithosphere gets oxidized to carbonate, thereby lowering the melting point and finally producing carbonated melts (Foley, 2011). Metasomatism commonly goes along with an increase of the oxidation state of the mantle (Rudnick et al., 1993; Rudnick et al., 1994a; Foley, 2011; Martin et al., 2017), which presumably is reflected by the high redox conditions for the rocks of the Kaiserstuhl. The lithological similarities between the Kaiserstuhl and many other carbonatite complexes indicate that high redox conditions are a common feature for carbonatite complexes worldwide. A metasomatically overprinted (sub)lithospheric mantle therefore might be an important prerequisite for the formation of such complexes (see also Study 7).

4.4 Multi-stage evolution of carbonatite magmas monitored by pyrochlore

Title of publication:

Pyrochlore as a monitor for magmatic and hydrothermal processes in carbonatites from the Kaiserstuhl volcanic complex (SW Germany). (Study 4)

Pyrochlore group minerals are common accessory phases in carbonatites and show significant variations in their chemical composition (e.g. Hogarth et al., 2000). Thus, they may be suitable to distinguish diverse carbonatite formation stages. In this study, we investigated pyrochlore in the sövite bodies, the extrusive carbonatites, and the alvikite and beforsite dykes of the Kaiserstuhl Volcanic Complex. Based on a textural, mineral chemical and geochemical analyses, we decipher the evolution of such a carbonatitic system. The whole-rock chemistry was determined using XRF (major elements) and ICP-MS (trace elements), while the mineral composition was determined by the electron microprobe (EPMA).

Four main pyrochlore types were distinguished at the Kaiserstuhl: (I) irregularly shaped pyrochlore cores that contain up to 22 wt.% UO_2 and 9 wt.% Ta_2O_5 ; (II) oscillatory-zoned pyrochlore rich in CaO (up to 23 wt.%) and Na_2O (up to 11 wt.%); (III) patchy pyrochlore zones that are characterized by high REE oxide concentrations of up to 18 wt.%, and which overgrew the before-mentioned pyrochlore types; (IV) unzoned pyrochlore crystals with high compositional variations between the different rock samples.

Irregularly shaped, U- and Ta-rich pyrochlore cores surrounded by an oscillatory-zoned pyrochlore generation occur in several carbonatites worldwide and are typically interpreted to represent the multi-stage magmatic evolution of a carbonatitic magma (Hogarth et al., 2000; Chakhmouradian and Williams, 2004; Lee et al., 2006). However, at the Kaiserstuhl, pyrochlore type I mainly occurs at the Badberg carbonatite pipe, which is characterized by cm- to m-sized silicate xenoliths (former nosean syenites). A xenocrystic origin of these U-Ta-rich pyrochlore cores is unlikely since pyrochlore could not be observed in the pristine nosean syenites. We rather suggest that the compositionally distinct pyrochlore generation formed by assimilation processes due to the (partly) resorption of the xenoliths (see also Study 5). This conforms with the

xenolith clasts of the inner part of the carbonatite pipe which are more strongly disaggregated than those of the peripheral zone.

The oscillatory zoned pyrochlore type II appears in all sövites and is considered a typical orthomagmatic feature, due to the faster crystal growth rate compared to the diffusion rate in tranquil conditions (Hogarth et al., 2000). The patchy pyrochlore zones (type III) mainly occur at the Orberg and illustrate a late magmatic to hydrothermal overprint. This is in agreement with the evolved character of these carbonatites, as demonstrated by the high whole-rock Nb/Ta and Zr/Hf ratios that could be caused by extensive fractionation of pyrochlore (Chakhmouradian, 2006). High REE concentrations in the whole-rock analyses and pyrochlore from Orberg also point towards an evolved magma composition.

Unzoned pyrochlore, restricted to the extrusive carbonatites, the alvikites and the beforsites, impedes a genetic discrimination. However, especially the tuffs at Henkenberg contain fragments of silicate rocks and may represent the volcanic equivalent to the sövites at the Badberg, which is also consistent with the similar pyrochlore composition in both carbonatite types.

4.5 Interaction processes between carbonatite magmas and silicate wall rocks

Title of publication:

Evidence for Magma–Wall Rock Interaction in Carbonatites from the Kaiserstuhl Volcanic Complex (Southwest Germany). (Study 5)

The low density and the low viscosity of carbonatitic melts enables them to ascend rapidly (Treiman and Schedl, 1983; Jones et al., 2013). Hence, even though the chemical contrast is large, contamination processes between carbonatites and crustal rocks are usually considered to play a negligibly role. However, if contamination by silicate rocks occurred, it is difficult to detect by means of radiogenic isotopes like Sr, Nd and Pb. Carbonatites generally have much higher concentrations of these elements, thereby buffering their isotope systems against crustal contamination. On the contrary, carbonatites offer considerable potential for magma-wall rock interactions due to their high geochemical contrast to most silicate rocks, their high liquidus temperature (1000 °C), and their high volatile contents (e.g. Treiman and Essene,

1984; Jago and Gittins, 1991; Vuorinen and Skelton, 2004; Chakhmouradian et al., 2008; Guzmics et al., 2011). Based on a textural and a mineral chemical investigation (EPMA) of the sövite bodies and the extrusive carbonatites, this study provides evidence for contamination processes in the carbonatitic magma system of the Kaiserstuhl Volcanic Complex.

Besides calcite, the carbonatites at the Kaiserstuhl typically consist of apatite, mica, spinel group minerals and pyrochlore. The extrusive carbonatites are characterized by nepheline, garnet and clinopyroxene. Olivine occurs in few sövites at Haselschacher Buck and at Katharinenberg, while small occurrences of monticellite-bearing carbonatites were found at the Orberg.

The sövites at the Badberg are peculiar since they contain euhedral clinopyroxene, high proportions of large mica crystals, and cm- to m-sized silicate xenoliths (calcite foidolites). These calcite foidolites largely consist of former nosean/haüyne (which has been completely altered to zeolite minerals), interstitial clinopyroxene, calcite, apatite, rare titanite and relict garnet which is surrounded by an interstitial garnet generation. Despite the textural similarity to the haüynolites, we suggest that the calcite foidolites represent the interaction product between the carbonatitic magma and the nosean syenites because of the chemical composition of the garnet relicts and the fact that the nosean syenites underlie the Badberg carbonatite (Kirchheimer, 1973). The contrasting mineral abundances and textures compared to the nosean syenites (higher foid content; lack of magnetite and alkali feldspar; interstitial clinopyroxene and garnet) demonstrates intense resorption and recrystallization processes.

The contact towards the calcite foidolite is characterized by a black wall of euhedral mica, which sometimes additionally comprise an inner zone of clinopyroxene, garnet and calcite. On a larger scale, similar observations can be drawn by the decrease of xenolith abundance, but the increase of mica content from the marginal towards the central part of the carbonatite body. This presumably is related to a more intense resorption of the nosean syenites due to a longer prevailing heat flow in the inner part of the carbonatitic magma pipe compared to the faster cooled pipe margins.

While the Badberg carbonatite is mica-rich and characterized by large mica grains up to 2 cm, all other carbonatites at the Kaiserstuhl are mica-poor and contain much smaller crystal sizes (< 2 mm). The micas at the Badberg are also chemically distinguishable from the other carbonatite types, as they contain comparably low

concentrations of Mg and Ba, but high contents of Mn and Fe. Especially the high proportion of Fe is atypical for mica in carbonatites (e.g. Lee et al., 2003; Krasnova et al., 2004a; Giebel et al., 2019), and implies an element supply by assimilation with the nosean syenite. Resorption of clinopyroxene, magnetite and alkali feldspar may have provided significant amounts of Fe, K, Al, and Si required for the crystallization of abundant mica.

The chemical composition of the apatites at the Badberg is also unique at the Kaiserstuhl. While the cores have a similar composition compared to the other carbonatite bodies, the rims demonstrate a relatively high britholite substitution ($\text{Ca}^{2+} + \text{P}^{5+} \leftrightarrow \text{REE}^{3+} + \text{Si}^{4+}$). The sharp contact between the core and the compositionally distinct apatite rim (rather than a continuous transition) argues against magmatic differentiation, although the comparably high Sr concentration of the Badberg apatites may reflect a high evolution stage of the carbonatite magma. However, a hydrothermal origin can be excluded since a high britholite component typically indicates magmatic crystallization conditions ($T > 600 \text{ }^\circ\text{C}$), whereas a hydrothermal overprint results in the crystallization of REE-poor apatite and separate REE-minerals (Anenburg et al., 2018; Anenburg and Mavrogenes, 2018). Thus, the high REE concentration in the apatite rims is the result of charge balancing due to the high incorporation of Si, which was provided by the (partly) resorption of the silicate xenoliths (Klemme and Dalpé, 2003; Hammouda et al., 2010). The enhanced incorporation of REE in early crystallizing apatite led to rather low REE concentrations in late-stage pyrochlore compared to the carbonatites without evidence of crustal contamination and/or assimilation. An elevated Si activity in the Badberg carbonatites is also indicated by the presence of euhedral clinopyroxene (Barker, 2001). The chemical composition is similar to the recrystallized, interstitial clinopyroxenes of the calcite foidolites, but differs from the clinopyroxenes of all other silicate rocks at the Kaiserstuhl (see also Study 3), hence ruling out a xenocrystic origin. This contrasts with garnet, clinopyroxene and some spinel group minerals of the extrusive carbonatites where the chemical composition indicates that the minerals represent xenocrysts entrapped from various silicate rocks. Similar to the Badberg, many carbonatites worldwide have high modal amounts of mica (e.g. Brod et al., 2001; Lee et al., 2003; Reguir et al., 2009; Giebel et al., 2019). However, experiments on carbonatitic melts demonstrate that the solubility for elements such as Si and Al is strictly limited (Brooker and Kjarsgaard, 2011; Weidendorfer et al., 2017). Based on mass balance calculations, we conclude that a

maximum of approximately 7 % mica can crystallize from a carbonatitic melt. Note that these calculations were examined by assuming that sufficient K, Mg and H₂O are available, and mica is the sole crystallizing mineral phase. We therefore suggest that the high mica content of many carbonatites can only be explained by a continuous interaction with silicate rocks and a consequently steady supply of elements such as Si and Al. The occurrence and composition of silicate minerals like mica and clinopyroxene hence may be a useful indicator for contamination and assimilation processes during the emplacement of carbonatitic magmas.

4.6 Interaction processes between silicate magmas/rocks and potash salt rocks

Title of publication:

Hydrothermal halite precipitation in mafic volcanic rocks: the case of the Buggingen salt deposit, SW Germany. (Study 6)

Olivine melilitites and monchiquites intruded into the potash salt deposit at Buggingen. Similar to the carbonatite magma-silicate wall rock interaction mentioned in Study 5, the large geochemical contrast between primitive dyke rocks and salt-rich host rocks provides potential for intense magmatic to hydrothermal assimilation processes. Based on a detailed mineralogical and mineral chemical characterization (EPMA), we reveal the influence of the wall rock on the mineralization of the dyke rocks.

The dyke rocks are characterized by spheroidal textures that show a contrasting mineralization depending on whether they adjoin the salt rocks or the bituminous Fish Shales. Roundish textures associated with the salt layer mainly comprise smectite, calcite, halite and sometimes anhydrite, while those associated with the Fish Shale consist of smectite, calcite, zeolite and analcime. As spheroidal textures are present in all investigated dyke rocks, independent of the host rock, we suggest that the halite inclusions do not illustrate salt-magma mingling, or salt-liquid immiscibility caused by intense magmatic assimilation with the salt rocks. The absence of salt rock-melt interactions might be related to a low orthomagmatic water content of the primitive melts, which inhibited intense fluid exsolution and a consequent dissolution of the salt. Moreover, the dyke magma presumably cooled too fast in order to melt the salt and

enable liquid mingling and/or assimilation processes (Knipping, 1989). The halite-bearing textures are interpreted as amygdules and hydrothermally altered ocelli. NaCl-bearing fluids derived from the salt layers reacted with the dyke rock groundmass and formed smectites and zeolites, thereby cementing the fluid paths and increasing the Na concentration of the fluid. The absence of halite in the groundmass indicates that the initial fluid was not saturated in NaCl. As soon as the fluid entered a vesicle or an ocellus where the calcite-bearing interior (Hurrell, 1973) was previously dissolved, smectite formed at the marginal areas and the fluid became trapped. Water extraction by clay mineral swelling finally led to the precipitation of halite. Datolite was observed in pseudomorphs after olivine, associated with halite or with serpentine/smectite. The required boron for the formation of datolite most likely derives from the clay-rich layers of the host rock (e.g. Leeman and Sisson, 1996; Garrett, 1998) and was incorporated/adsorbed into serpentine/smectite during alteration of olivine. Subsequent breakdown of serpentine/smectite during interaction with a second fluid batch provided Si and B for the formation of datolite.

4.7 The relationship between the redox conditions of alkaline silicate rocks, carbonatites, and the mantle source

Title of publication:

Do carbonatites and alkaline rocks reflect variable redox conditions in their upper mantle source? (Study 7)

In our first study on the Kaiserstuhl (Study 3), we conclude that the high redox estimates for the various rock types may be a common characteristic for carbonatite complexes and suggest that the rocks may reflect a high oxidation state of their mantle source. This assumption was tested by a worldwide compilation of the intrinsic parameters of alkaline rocks and carbonatites, including new results from the carbonatite complexes Sokli, Kovdor, Palabora, Magnet Cove, Oka and Jacupiranga.

The chemical composition of the minerals was determined using the electron microprobe (EPMA). Based on these results, the formation conditions (temperature and oxygen fugacity) for the carbonatites (including phoscorites) and silicate rocks were calculated by several thermo(oxybaro)meters and thermodynamically stable

mineral equilibria. Temperatures typically vary between 900 and 500 °C, while the redox calculations indicate relatively oxidized conditions ($\Delta\text{FMQ}\approx 0$ to +3) for all rock types.

Including literature data, the reported crystallization temperature for carbonatites strongly depends on the applied geothermometer and ranges from ~1000 °C to less than 200 °C. The highest values correspond to calculations based on melt inclusions (e.g. Guzmics et al., 2011), while the lowest values can be explained by late magmatic to post-magmatic diffusion or exsolution processes (e.g. Haynes et al., 2003; Milani et al., 2016). The transition from magmatic to hydrothermal/carbothermal conditions can be determined by fluid inclusions which illustrate homogenization temperatures of up to 600 °C (Rankin, 1975). This is in broad accordance with experimental constrains of the carbonatite eutectic (~650 °C; Lee et al., 2000) and with melt inclusions in late-stage carbonate crystals (530 °C; Sokolov, 2014). We therefore conclude that magmatic carbonatites typically crystallize between ~1000 and 500 °C. To minimize the potential effect of fractional crystallization or thermal re-equilibration, we only consider redox estimates for temperatures above 600 °C.

Based on a worldwide compilation, carbonatites ($\Delta\text{FMQ}=+1.3$) and their associated alkaline silicate rocks at continental settings ($\Delta\text{FMQ}=+1.4$) formed on average under significantly higher redox conditions than alkaline silicate rocks at continental settings which are not associated with carbonatites ($\Delta\text{FMQ}=-1.1$). We suggest that these redox variations reflect the heterogeneity of their mantle source, although some variations may result from variable melting degrees, fractionation of Fe-bearing minerals and degassing processes (Carmichael, 1991; Gaillard et al., 2015). The prerequisite of a carbonated and relatively oxidized mantle for the formation of carbonatite complexes conforms with the high redox estimates for mantle xenoliths that show evidence for metasomatism processes involving carbonate-rich liquids (up to $\Delta\text{FMQ}=+3.5$; Rudnick et al., 1994b; Woodland et al., 1996; Uenver-Thiele et al., 2017). Such xenoliths may not only represent the interaction between the lithospheric mantle and ascending (carbonate-bearing) melts/fluids but may reflect a possible mantle source for the formation of carbonatite complexes as well. Since carbonatites are typically associated with nephelinitic to melililitic rocks, the source region for most carbonatite complexes can be estimated to depths of 75 to 130 km (Wyllie et al., 1990; Hirose and Kushiro, 1993; Gudfinnsson and Presnall, 2005). At such depths, the stability of carbonate

requires redox conditions above that of the average asthenospheric and lithospheric mantle (Foley, 2011), roughly in the same range than those for carbonatites and associated silicate rocks.

Carbonatites and alkaline silicate rocks at the seven investigated complexes all indicate oxidized formation conditions. However, the various rock types do not always originate from the same mantle source (e.g. Kaiserstuhl, see Study 3), hence demonstrating that oxidation processes must have affected multiple mantle domains. An increasing redox state of the mantle may result from subduction and digestion of oxidized (and eventually carbonate-bearing) crustal material (Yaxley and Brey, 2004; Foley, 2011; Rohrbach and Schmidt, 2011), or by the interaction with ascending asthenospheric liquids. Note that partial melts presumably have a higher oxidation state than the precursor solid (Canil and O'Neill, 1996; Frost and McCammon, 2008; Mallmann and O'Neill, 2009).

A rather high oxidation state and enough carbonate in the mantle, however, is not the sole prerequisite for the formation of carbonatite complexes. The mantle beneath subduction zones, for example, is relatively oxidized, but carbonatites are absent presumably due to the reversed temperature gradient of the mantle wedge. Ascending carbonate-bearing melts would react with the hot overlying mantle, whereby they would be strongly modified and diluted (Yaxley et al., 2019). Ocean island basalts and spatially associated xenoliths also indicate oxidized conditions (Ballhaus, 1993; Mallmann and O'Neill, 2007; Foley, 2011). Such environments typically lack carbonatites because of a too thin lithosphere which prevents the parental melts to evolve into the carbonatite-silicate miscibility gap (Weidendorfer et al., 2017). Further reasons for the absence of carbonatites at alkaline complexes that formed under oxidized conditions include CO₂ degassing processes, the complete crystallization of the silicate melt before an exsolved/residual carbonatite magma could form, or lack of outcrop.

5 Conclusion and implications

This thesis provides new and improved insights into the crystallization conditions and processes of variably evolved alkaline silicate rocks and carbonatites. It was shown that the rock type and the mineral assemblage may strongly vary as a function of the mantle source conditions, magma mixing, and crustal contamination/assimilation processes. Especially alkaline complexes that comprise agpaitic rocks reflect contrasting crystallization conditions and magma evolution processes compared to alkaline complexes associated with carbonatites. This can be impressively demonstrated by the Tarosero Volcanic Complex (Tanzania) and the Kaiserstuhl Volcanic Complex (Germany).

In general, the main prerequisite for the formation of agpaitic rocks is the retention of volatiles in the melt. Thus, almost all agpaitic occurrences represent intrusive rocks. In case of the extrusive agpaitic rocks at Tarosero, the near-liquidus crystallization of OH-bearing amphibole resulted in a low water activity at late-magmatic conditions which prevented the exsolution of fluids and water-soluble halogens. Consequently, the solubility of REE and other HFSE was kept high in the melt residue, which finally supported the formation of interstitial eudialyte. The rocks from Tarosero derive from a basaltic magma source and evolved by extensive fractional crystallization at low redox conditions ($\Delta\text{FMQ} \leq 0$).

This stands in great contrast to the rocks of the carbonatite-bearing Kaiserstuhl volcano. A K-basanitic magma produced the tephritic to phonolitic rock series, while an olivine melilititic magma is responsible for the nephelinitic to limburgitic rock series, the melilititic to haüynitic rock series and the carbonatites. Fractional crystallization at high redox conditions ($\Delta\text{FMQ} = +1$ to $+2$) is the key process for the evolution of such magmas.

In fact, carbonatites and associated silicate rocks at continental settings reveal on average significantly higher redox conditions ($\Delta\text{FMQ} \approx +1.4$) than alkaline complexes without carbonatites ($\Delta\text{FMQ} = -1.1$). Especially agpaitic rocks typically demonstrate reduced formation conditions (Marks and Markl, 2017). The redox variations presumably reflect the heterogeneity of the oxidation state of the mantle source. This is consistent with the prerequisite of a relatively oxidized, carbonate-bearing mantle for the formation of carbonatites, and may explain why carbonatites are absent at most agpaitic complexes. However, there are few exceptions where carbonatites are

spatially associated with agpaitic rocks (e.g. Magnet Cove, Oldoinyo Lengai, Khibina, Kovdor, Tamazeght, Kontozero, Kaiserstuhl; Erickson and Blade, 1963; Dawson and Frisch, 1971; Kramm and Kogarko, 1994; Pekov et al., 2001; Marks et al., 2008; Petrovsky et al., 2012; Marks and Markl, 2017). It remains doubtful that these agpaitic occurrences all derive from separate, relatively reduced mantle sources. Indeed, agpaitic rocks can also derive from primitive nephelinitic magmas (Marks et al., 2011; Schilling et al., 2011), which may also be parental to carbonatites. However, redox calculations for presumably nephelinite-derived agpaitic rocks are restricted to the Tamazeght Carbonatite Complex and “have to be taken as rough estimates” (Marks et al., 2008). Further investigations on such rocks are needed to make generally valid statements about their redox conditions. Moreover, there are several further processes that have to be considered to relate the oxidation state of the mantle with its magmatic derivatives. For example, we demonstrated evidence for multi-stage magma mixing in the Ogol basalts of the Laetoli area. Mixing of several magmas with contrasting redox conditions, however, may result in an intermediate oxidation state, which does not reflect the redox conditions of the mantle source. At the Kaiserstuhl, we showed that carbonatite magma-wall rock interactions caused an increasing silica activity of the magma, and it seems likely that such processes may also affect the redox conditions. Variable melting degrees, degassing and fractional crystallization may also influence the oxidation state of a magmatic system (Gaillard et al., 2015). Thus, further work on such processes are appreciated in order to more precisely link the redox conditions of igneous rocks to their mantle source.

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Appendix I

Submitted manuscript

Study 1

Zaitsev, A. N., Arzamastsev, A. A., Marks, M. A. W., Braunger S., Wenzel, T., Spratt, J., Salge, T., and Markl, G., submitted. Multiple sources for basaltic volcanism in Laetoli area (Crater Highlands, Tanzania): evidence from mineralogy and geochemistry of the Ogol lavas. *Journal of Petrology*.

1 **Multiple sources for basaltic volcanism in Laetoli area (Crater Highlands, Tanzania): evidence**
2 **from mineralogy and geochemistry of the Ogol lavas**

3

4 Antoly N. Zaitsev^{1,2*}, Andrei A. Arzamastsev^{3,4}, Michael A.W. Marks⁵, Simon Braunger⁵, Thomas
5 Wenzel⁵, John Spratt², Tobias Salge² and Gregor Markl⁵

6

7 ¹Department of Mineralogy, St. Petersburg State University, University Emb. 7/9, St. Petersburg,
8 199034, Russia; ²Imaging and Analysis Centre, Department of Earth Sciences, Natural History
9 Museum, Cromwell Road, London, SW7 5BD, UK; ³Department of Petrography, St. Petersburg State
10 University, University Emb. 7/9, St. Petersburg, 199034, Russia; ⁴Institute of Precambrian Geology
11 and Geochronology, Russian Academy of Sciences, Makarova Emb. 2, St. Petersburg, 199034,
12 Russia; ⁵Universität Tübingen, Mathematisch-Naturwissenschaftliche Fakultät, FB
13 Geowissenschaften, Wilhelmstrasse 56, D-72074 Tübingen, Germany

14

15 *Corresponding author. E-mail: a.zaitsev@spbu.ru

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17 Running title: Ogol basalts

18

19 **ABSTRACT**

20

21 The southern part of the eastern branch of the East African Rift is characterized by extensive magmatism
22 since the late Miocene. In the Crater Highlands, part of the North Tanzanian Divergence zone, effusive
23 and pyroclastic rocks reflect nephelinitic and basaltic compositions that formed between 4.6 and 0.8 Ma.
24 The former are best represented by the Sadiman volcano (4.6-4.0 Ma) and the latter occur in the giant
25 Ngorongoro crater (2.3-2.0 Ma), the Lemagarut volcano (2.4-2.2 Ma) and as a small volcanic field in
26 the Laetoli area (2.3 Ma), where basaltic rocks known as Ogol lavas were erupted through fissures and
27 several cinder cones. Compositionally, they are alkaline basalts with 46.0-47.9 wt % SiO₂, 3.0-4.3 wt %
28 of Na₂O+K₂O, Mg# of 61 to 55, and high Cr and Ni content (450-975 and 165-222 ppm respectively).
29 A detailed textural and compositional analysis of the major minerals (olivine, clinopyroxene, spinel-
30 group minerals, plagioclase) in the rocks reveals their heterogeneity. Most samples contain appreciable
31 proportions of xenocrystic minerals, with large variations between samples. The primary mineral
32 assemblage that crystallized from the Ogol magmas comprises forsterite (Fo_{89.5-84.2}), Cr-bearing diopside
33 to augite, magnesiochromite-chromite, magnetite-ulvöspinel, andesine-oligoclase and apatite, with glass
34 of phonolitic composition in the groundmass. Xenocrysts that derive from distinct sources include
35 forsterite (Fo_{84.1-72.5}) with reverse zonation, rounded and embayed clinopyroxene cores of variable
36 composition (at least four different populations), Cr-free magnetite-ulvöspinel and oligoclase. These
37 xenocrysts as well as observed variations in major and trace element contents, variable ⁸⁷Sr/⁸⁶Sr_(i)
38 (0.70377-0.70470) and ¹⁴³Nd/¹⁴⁴Nd_(i) (0.51246-0.51261) isotopic ratios provide evidence of multi-stage
39 magma mixing between Ogol and Lemagarut basaltic melts. Elevated alkalinity, which positively
40 correlates with isotope ratios and the presence of xenocrystic perovskite and schorlomite, indicate
41 additional mixing with alkaline melts and/or assimilation of alkaline rocks (related to the Sadiman
42 volcano and/or the Laetoli tuffs). Contamination by crustal granite-gneisses did not exceed 5 %. Due
43 to their heterogeneity, estimates on the crystallization conditions for the Ogol rocks are difficult.
44 Nevertheless, clinopyroxene-liquid thermobarometry indicates crystallization temperatures of around
45 1150-1220 °C and records upper-crustal depths of 3-12 km (1-4 kbar). Despite the fact that Ogol basalts

46 are hybrid rocks that formed under open-system conditions, they seems to be the best examples of near-
47 primary basaltic melts within the Crater Highlands.

48

49 **Key words:** basalt; Crater Highlands; Laetoli; magma mixing; xenocryst

50

51 INTRODUCTION

52

53 The eastern branch of the East African Rift (EAR) that opened along the border of Tanzanian craton is
54 known as a region of intense magmatism developed since the late Miocene to Recent. Numerous
55 volcanic centers in the North Tanzanian Divergence zone (NTD; Fig. 1; Baker *et al.*, 1972) produced a
56 wide variety of lavas and pyroclastic rocks ranging in composition from undersaturated peralkaline
57 rocks and carbonatites to subalkaline basalts. Models for the structure of the NTD and the sequence of
58 magmatic events directly depend on the accuracy of geochronological data, most of which were obtained
59 in the 1960s and 1970s using low-precision methods, e.g. whole-rock K-Ar data (reviewed in Dawson,
60 2008). Recent $^{40}\text{Ar}/^{39}\text{Ar}$ ages allow to decipher the mixture of undersaturated peralkaline and basic rocks
61 forming eruptive centers of different age and discriminate distinct pulses of igneous activity in the NTD
62 (Mollel, 2007; Nonnotte *et al.*, 2008; Mana *et al.*, 2012, 2015; Peirce & Mana, 2018).

63 The main source of magmatism in the NTD area is probably the metasomatized subcontinental
64 lithospheric mantle (Mana *et al.*, 2015), but the composition of potential primary magmas remains
65 unclear. Recent nephelinite-phonolite volcanic activity, which produced several centers along the axis
66 of Gregory rift in Lake Natron - Engaruka area may relate to olivine melilitite primary melts (Keller *et*
67 *al.*, 2006), However, the basalt-trachybasalt-trachyandesite-trachyte series documented from
68 Lemagarut, Ngorongoro and Oldeani volcanoes and the Olduvai Gorge can not be easily explained by
69 fractionation of such parental melts. Basaltic rocks known as Ogol lavas, forming lava flows in the
70 Laetoli area, 30 km west of the Ngorongoro caldera (Mollel, 2007), have characteristics that allow them
71 to be classified as primary magmas of basalt-trachyte series. In the present study we use the
72 geochemistry and mineralogy of a suite of mafic Ogol lavas to constrain their origin, their depth of
73 melting and source region character. We integrate these results with evolutionary paths of basalt-trachyte

74 series in adjacent volcanoes and evaluate the role of possible crustal contamination and magma mixing.

75

76 **GEOLOGICAL SETTING**

77

78 **Regional geology**

79 The Crater Highlands and adjacent Gregory Rift are parts of a much larger volcanic province, extending
80 from the Afar triple junction southwards to the Mozambique margin along the western and eastern
81 borders of the Tanzanian Craton, including numerous complexes in Ethiopia, Kenya and Tanzania. In
82 the NTD zone in the southern end of the southerly-propagating eastern branch of the EAR system, the
83 distribution of volcanoes is controlled by lineaments and faults along the boundary of the Neoproterozoic
84 Tanzanian Craton and the Proterozoic Mozambique belt (Smith, 1994) (Fig. 1).

85 The earliest manifestation of volcanic activity is represented by an alkali picrite-nephelinite-
86 phonolite series from Essimingor volcano dated to 5.91 ± 0.02 Ma (Mana *et al.*, 2012). Further alkaline
87 magma eruptions throughout the NTD zone formed Sadiman, Mosonik, Engelosin, Tarosero and other
88 minor centers ranging in age from 4.6 to 2.3 Ma (Dawson, 2008). Silica undersaturated alkaline magma
89 eruptions were replaced and superimposed by Lemagarut, Ngorongoro, Monduli, Olduvai, Olmoti,
90 Oldeani volcanoes, calderas and minor centers in which basalts and basanites are the predominant rock
91 types. Vast basalt lava beds spread along the southern end of Gregory Rift in the Engaruka-Manyara
92 area.

93 Based on $^{40}\text{Ar}/^{39}\text{Ar}$ age determinations basaltic magmatism lasted from 2.4 Ma to 1.6 Ma
94 (Manega, 1993; Foster *et al.*, 1997; Mollel, 2007). Since 1.3 Ma renewal of magmatism was marked by
95 ongoing activity in the axial part of NTD, and its lateral eastward spreading. Alkaline rocks ranging in
96 composition from primitive olivine melilitite to highly evolved phonolite, sporadically associated with
97 carbonatite from the Embagai, Burko, Kerimasi, Meru, and Oldoinyo Lengai volcanoes and numerous
98 minor craters and cones in the Lake Natron-Engaruka area (Dawson, 2008; Mattsson *et al.*, 2013).

99 In multiphase volcanic centers, such as Kilimanjaro and Ketumbeine, early mafic magmatism
100 started at 2.2 Ma (simultaneously with Ngorongoro and other basaltic centers) followed by phonolitic
101 eruptions dated to 1.3 and 0.5 Ma, respectively (Nonnotte *et al.*, 2008; Peirce & Mana, 2018). A recent

102 study (Mana *et al.*, 2015) provides evidence of consistent younging of NTD magmatism along two main
103 trends: one from west to east (Essimingor to Kilimanjaro) and another from SW to NE (Sadiman to
104 Gelai).

105

106 **Crater Highlands**

107 The Crater Highlands comprise numerous eruptive centers in a 100 km-long elevated area between lakes
108 Natron and Eyasi at the southern termination of the eastern branch of the EAR (Fig. 2). Seismic
109 tomography reveals a thick mantle keel (exceeding 200 km) beneath the craton (Ritsema *et al.*, 1998;
110 Adams *et al.*, 2012), considerably thicker than the lithosphere beneath the Eastern Rift north of Lake
111 Natron, which is less than 100 km thick (Birt *et al.*, 1997). Although the contact between Archaean and
112 Late Proterozoic crust is west of the Crater Highlands, outcrops of gneissic-granitic basement indicate
113 that Archaean lithosphere underlies much of this rift sector. Recent local magnetotelluric and
114 tomography studies (Plasman *et al.*, 2019) indicate the presence of a melt accumulation zone at 40 km
115 depth below the Crater Highlands, potentially connected to the volcanic edifices at the surface. The
116 Eyasi fault system, which delimits the westward distribution of magmatism in upland area, is interpreted
117 as a narrow vertical conduit linking the surface to a deeper conductive structure between two large
118 basement terrains (Plasman *et al.*, 2019).

119 The earliest volcanic activity, beginning at 4.63 ± 0.05 Ma (Mollet *et al.*, 2011), produced large
120 volumes of interlayered phonolitic tuffs, tuff breccias and nephelinitic lava flows erupted from Sadiman
121 volcano (Zaitsev *et al.*, 2011; 2012). Superimposed tuffs west of Sadiman, in particular the Laetolil
122 Footprint Tuff, where 3.66 Ma *Australopithecus afarensis* footprints were discovered (Leakey & Hay,
123 1979), may have been erupted from Mosonik volcano (Zaitsev *et al.*, 2015; 2019). Since about 2.4 Ma,
124 mafic magmatism predominates in the Crater Highlands. Eruptions from the Lemagarut volcano
125 occurred between 2.40 ± 0.01 Ma and 2.22 ± 0.10 Ma and the giant Ngorongoro volcano was active
126 between 2.25 ± 0.02 Ma to 2.01 ± 0.02 Ma (Mollet *et al.*, 2008; 2011), erupting basalts, basanites and their
127 evolved derivatives (trachyandesite, trachydacite and trachyte). The youngest manifestations of mafic
128 magmatism is represented by the Oldeani volcano which was active between 1.61 ± 0.01 Ma and
129 1.52 ± 0.02 Ma (Mollet, 2007).

130 **Ogol lavas**

131 Basaltic lavas (local name Ogol lavas) erupted through fissures and cinder cones ranging from 0.5 to
132 1.5 km in diameter and 50 to 150 m in height (Hay, 1987; Supplementary Data Fig. S1). The major
133 localities (as named on Quarter degree sheet 52 Endulen; Pickering, 1964) are Oldoinyo Mati
134 (Ndonyamati), Oldoinyo Emusenge (Norsigidok), Oldoinyo Ildolaunya (Ndolanya) and Oldoinyo
135 Naisiusiu hills (Fig. 3). Outcrops with well exposed lava flows are limited, but rounded lava blocks up
136 to 2.5 m in diameter are widespread (Supplementary Data Fig. S1). The Ogol lavas are massive
137 (common) to vesicular (rare), fine- to medium-grained, and have aphanitic (rare) to porphyritic
138 (common) textures. The proportions of phenocrysts (olivine and pyroxene) are highly variable and range
139 from approximately 1 to 30 vol. % (Supplementary Data Fig. S2). The Ogol lavas overlie the Ndolanya
140 (3.58-2.66 Ma) and Laetolil (4.36-3.66 Ma) beds and are partly overlain by the Naibadad beds (2.15-
141 2.06 Ma) (Hay, 1987; Dawson, 2008; Deino, 2011). The eastern part of the flows is superimposed by
142 basalts and basaltic andesites of Lemagarut volcano. Mollel (2007) reported two $^{40}\text{Ar}/^{39}\text{Ar}$ ages
143 (2.31 ± 0.01 Ma and 2.27 ± 0.05 Ma) collected from the northeast trending Ogol lava flow west of
144 Lemagarut volcano.

145 Despite their importance for the understanding of basaltic volcanism in the Crater Highlands
146 (Mollel, 2007) as well as in anthropological studies in Laetoli (Adelsberger *et al.*, 2011), the Ogol lavas
147 are poorly studied. They were originally described as vogesites (Hay, 1987) and subsequently classified
148 as basalts (Mg#=49-59) with olivine and augite as phenocrysts, and plagioclase, augite and Ti-rich
149 magnetite as groundmass phases (Mollel, 2007). Although, this study provided high precision ages (see
150 above), geochemical and mineralogical data are limited and incomplete. A more extensive geochemical
151 study was published by Adelsberger *et al.* (2011). Based on whole-rock data, the rocks were classified
152 as basalts and basanites with each one sample of tephrite and picrobasalt. The basaltic and some of the
153 basanitic samples are relatively primitive (Mg# of 55-61) with appreciable Cr (496-1091 ppm) and Ni
154 (188-248 ppm). Three basanitic samples are more evolved, Mg#=47-48, with 297-338 ppm Cr and 77-
155 95 ppm Ni. A picrobasalt is strongly enriched in Fe and depleted in Mg (Mg# of 20) and is low in Cr
156 (32 ppm) and Ni (8 ppm) and unusually high P (2.1 wt % P_2O_5).

157

158 **SAMPLES AND METHODS**

159

160 For this study we investigated 15 samples of Ogol lavas and 12 samples of lavas from the western part
161 of the Lemagarut volcano (Fig. 3). Crystalline basement rocks are not exposed in the Laetoli area, but
162 at the Eyasi plateau about 15 km south and south-west from the Laetoli area. We collected 8 samples
163 from basement rocks of the Tanzania craton, comprising granite-gneisses, green and melanocratic
164 schists and garnet amphibolites with an age of about 2.7 Ga (Supplementary Data Table S1; Pickering,
165 1964; Fletcher *et al.*, 2018).

166 Mineral identification was based on petrographic studies of polished thin sections (Leica DM
167 2500P microscope and Hitachi S-3400N scanning electron microscope, St. Petersburg State University)
168 and XRD measurements of powder samples using a Bruker D8 Focus XRD system (St. Petersburg State
169 University).

170 Whole-rock samples (lavas and basement rocks) were analysed for major and trace elements by
171 X-ray fluorescence (XRF) and inductively coupled plasma mass-spectrometry (ICP-MS) at the Karelian
172 Research Centre RAS (Petrozavodsk). Powders were fused into glass disks using ca. 3 g of sample
173 mixed with lithium tetraborate ($\text{Li}_2\text{B}_4\text{O}_7$). Glass disks were analyzed by XRF using an ARL
174 ADVANT"X X-ray sequential fluorescence spectrometer. Loss on ignition (LOI) was determined
175 gravimetrically. The uncertainties were 1–5% for elements with concentrations of >0.5 wt % and up to
176 12% for elements with concentrations of <0.5 wt %. The concentrations of trace elements were
177 determined by ICP-MS using an X Series 2 instrument. Powdered samples were digested in acid mixture
178 in open vessels following the standard procedure described by Svetov *et al.* (2015). The accuracy of the
179 analyses was monitored by analyzing USGS reference materials BHVO-2 and BIR-1a, and in-house
180 1412 and SGD-2a reference materials. Precision is generally < 1% for Sr, Y, Zr, Nb, Cr, Ni, Cu, Zn, V,
181 La, Ce, Pr, Nd; < 3% for Ba, Sr, Rb, Cs, Sc, Co, Sm, Eu, Gd, Tb, Dy, Ho, Er, Hf, Ta, Yb, Lu, Pb Th
182 and U.

183 Mineral compositions (olivine, clinopyroxene, spinel and plagioclase) were obtained by electron
184 microprobe analysis (wavelength-dispersive X-ray spectrometry mode) using a JEOL 8900 Superprobe
185 (Tübingen University) and Cameca SX-100 (Natural History Museum, London) instruments operated

186 at same analytical conditions (20 kV, 20 nA with a beam diameter of 1 μm for olivine, clinopyroxene
187 and spinel, and 10 μm for plagioclase). Well-characterized natural and synthetic compounds were used
188 for calibration. Glass compositions were obtained by using a Hitachi S-3400N scanning electron
189 microscope (St. Petersburg State University) equipped with an Oxford Instruments X-Max 20
190 spectrometer (20 kV, 1.0 nA and defocused beam 10-15 μm in diameter).

191 The Rb-Sr and Sm-Nd isotopic compositions of the 10 Ogol samples, 3 Lemagarut samples, and
192 3 basement rocks samples were analysed using a multi-collector TRITON mass-spectrometer operated
193 in static mode (VSEGEL, St. Petersburg). Detailed information on sample preparation and analysis
194 procedure is given in Zaitsev *et al.* (2019).

195

196 **WHOLE-ROCK GEOCHEMISTRY**

197

198 Whole-rock geochemical data for 13 Ogol samples are given in Table 1. The studied samples have
199 relatively low SiO_2 (46.0-47.9 wt %) and $\text{Na}_2\text{O}+\text{K}_2\text{O}$ (3.0-4.3 wt %) contents, and are rich in MgO (8.8-
200 11.7 wt %), CaO (10.0-12.3 wt %) and $\text{Fe}_2\text{O}_{3\text{total}}$ (13.2-15.5 wt %), with Mg# values (calculated as
201 $100 \cdot \text{Mg}/(\text{Mg}+0.9\text{Fe}_{\text{total}})$) ranging from 61 to 55. In a Total Alkali vs SiO_2 diagram (Le Bas *et al.*, 1986)
202 most Ogol samples plot in the alkaline basalt field (Fig. 4a). For further comparison and discussion, we
203 use published (Nonnotte, 2007; Mollel, 2007; Mollel *et al.*, 2008; Adelsberger *et al.*, 2011) and our
204 unpublished data from adjacent Lemagarut volcano and Ngorongoro caldera that are relevant to
205 evaluating 2.4-2.2 Ma magmatism in the Crater Highlands. Data from the nearby located but older
206 nephelinitic Sadiman volcano (Zaitsev *et al.*, 2012) are also considered as they are important for future
207 discussion on mineralogy of the basalts (see below).

208 The Ogol basalts are not compositionally uniform, particularly a suite of three samples is high
209 in TiO_2 (4.1-4.2 wt %), low in MgO (7.6-8.0 wt %) and low in SiO_2 (45.0-45.6 wt %) (Adelsberger *et al.*
210 *et al.*, 2011). Overall, there are negative correlations between the contents of MgO and SiO_2 , Al_2O_3 , Na_2O ,
211 K_2O and P_2O_5 (Fig. 4b, c) and less evident positive correlations of MgO with Fe_2O_3 and CaO. The
212 contents of compatible elements, such as Cr and Ni, are variable and high, with 297-1091 ppm and 77-
213 248 ppm, respectively (Fig. 4d). The lowest Cr (297-338 ppm) and Ni (77-95 ppm) contents are observed

214 in high Ti and low Mg-Si basalt (Adelsberger *et al.*, 2011).

215 The Ogol lavas are broadly enriched in incompatible trace elements with respect to the primitive
216 mantle and are characterized by slightly positive Ba and Pb anomalies and negative U, K and P
217 anomalies (Fig. 5a), some samples display marked positive Zr and Hf anomalies. The contents of rare
218 earth elements (REE) range from 206 to 276 ppm (Table 1) and there is a tendency of increasing REE
219 content with decreasing MgO. Chondrite-normalized REE patterns are enriched in light REE (LREE)
220 with $(La/Yb)_N$ ratios of 17.8-25.0 and $(La/Lu)_N$ ratios of 20.5-27.2 (Fig. 5b).

221 Considerable scatter in the major and trace element data of the studied samples argue against
222 simple closed-system fractional crystallization or mixing between two hypothetical endmembers. The
223 reasons for that will be understandable based on the following detailed petrographic descriptions and
224 mineral chemical investigations.

225

226 **PETROGRAPHY AND MINERAL CHEMISTRY**

227

228 The studied Ogol samples are characterized by variable structures, textures and proportions of mineral
229 crystals (Supplementary Data Fig. S2). In this study we prefer not to use the term “pheno- and
230 microphenocrysts”, because, as recent studied showed, phenocrysts in volcanic rocks commonly
231 represent very complex mineral assemblage. They consists of autocrysts (or phenocrysts *sensu stricto*),
232 primary minerals that are cogenetic with their host rock; antecrysts, crystals that crystallized from the
233 same magmatic system, but have been recycled one or several times before inclusion in the specific
234 sample; microlites, crystallized during magma degassing magma on eruption, and xenocrysts, i.e. these
235 crystals were entrained from other sources (e.g. Jerram & Martin, 2008; Ubide *et al.*, 2014; Shane *et al*
236 2019). Therefore, the terms macro- and microcrysts will be used for petrographic description of the Ogol
237 basalts. The only macrocrysts (>500 μm) are olivine and clinopyroxene, with the latter being commonly
238 more abundant (Fig. 6). They are generally 1-3 (rarely up to 5) mm in size, few of them reaching
239 megacrystic size of 9 x 3 mm (olivine) and 9 x 8 mm (clinopyroxene). Microcrysts (100-500 μm) include
240 olivine, clinopyroxene and rare plagioclase and spinel (not spinel *s.s.*, but spinel as a mineral group).
241 Groundmass minerals (<100 μm) are olivine (rare), clinopyroxene, plagioclase and spinel. Textural data

242 suggest the presence of several populations of olivine, clinopyroxene and spinel. Macro-, microcrysts
243 and groundmass crystals are characterized by different morphologies, zonation patterns, internal
244 structures and compositions.

245 Mosaic mapping of thin sections and blocks showed that apatite is relatively abundant in the
246 groundmass, and rare pyrrhotite and chalcopyrite occurs as inclusions in clinopyroxene. Previously,
247 these minerals have not been found in the Ogol basalts (Mollel, 2007; Adelsberger *et al.*, 2011). Ilmenite
248 was mentioned by Adelsberger *et al.* (2011) as a groundmass mineral but was not identified in our
249 samples as a primary mineral. As Adelsberger *et al.* (2011) did not describe (abundant) spinel in Ogol
250 basalts, we suggest that they may have mixed up ilmenite with spinel.

251 The majority of the studied samples show variable degrees of alteration and the most altered
252 samples (A16-24 and A16-25) derive from Oldoinyo Ildolanya hill. Olivine is partly to fully replaced
253 by fine-grained brown hydrous Mg, Fe-rich silicate (not precisely identified); calcite is a common
254 secondary mineral and fills vesicles and forms veins and veinlets. SEM/EDS studies also indicate the
255 presence of a montmorillonite, sodalite, Na-K feldspar and Na (+/-K) zeolite. This mineral assemblage
256 is considered to be a result of glass alteration. Further secondary minerals include Sr-bearing baryte
257 (veinlets and small nests), and rare Ba-bearing celestine, Ba-Mn hydroxide and dolomite (Fig. 6b).
258 Exceptions are two samples from the Oldoinyo Emusenge hill (A16-05 and A16-06) that show only
259 little alteration (Fig. 6a).

260

261 **Olivine**

262 Most olivine macrocrysts (and some of the larger microcrysts) are “hopper crystals” (Fig. 7) but they
263 also occur as euhedral to subhedral crystals, partly showing initial stages of skeletal growth at crystal
264 edges (Fig. 7b, e). Few grains are angular to sub-angular (Fig. 7g, f) and their morphology suggest that
265 they represent fragments of broken crystals, showing a kink-banded texture. The overall compositional
266 range (311 analyses from 10 samples) is Fo_{89.5} to Fo_{72.5} (Table 2, Fig. 8) and based on their internal
267 structures and compositions, we distinguish four forsterite populations with

268 (1) Normal zonation (Mg-rich core with Fe-rich rim, macro- and rare microcrysts),

269 (2) Reverse zonation (Fe-rich core with Mg-rich rim, macro- and extremely rare microcrysts),

270 (3) Repetitive zonation (Fe-rich core, Mg-riched mantle and Fe-rich rim, extremely rare
271 macrocrysts),

272 (4) No zonation (majority of microcrysts and groundmass crystals, rare crystal fragments of
273 various size).

274 The proportion between normally and reversely zoned crystals is highly variable. In samples
275 from Oldoinyo Emusenge (A16-05 and A16-06) reversely zoned forsterite is extremely rare, in other
276 samples normally zoned and reversely zoned forsterite occur in similar amounts, and in sample A19-64
277 reversely zoned forsterite dominates. Forsterite grains showing repetitive zonation are extremely rare,
278 with only two crystals observed in one sample (A19-01). Angular to sub-angular grains (unzoned or
279 with partial rim) are less common, but abundant in a sample from Oldoinyo Naisiusiu (A19-08).

280 Normally zoned forsterite has a core composition with $Fe_{89.5}$ to $Fe_{84.2}$ with NiO between 0.44
281 and 0.20 wt %. Both Fo and Ni contents decrease towards the rim with $Fe_{85.9-80.7}$ and 0.31-0.17 wt %
282 NiO. In reversely zoned forsterite, the cores are variable in composition ($Fe_{72.5}$ to $Fe_{84.1}$; 0.05-0.31 wt
283 % NiO) and their rims are always higher in Fo ($Fe_{80.2-86.0}$) and NiO (0.18-0.32 wt %). A single forsterite
284 megacryst (sample A19-08) is compositionally similar to the cores in reversely zoned forsterite, but it
285 is normally zoned with a Fo and Ni rich core ($Fe_{79.2-77.6}$, 0.22-0.17 wt % NiO) and a Fo and Ni depleted
286 rim ($Fe_{77.3-76.4}$; 0.15-0.14 wt % NiO). Microcrysts and groundmass forsterite has a composition similar
287 to that of rim on forsterite crystals with both zonation types. Aphanitic samples (A19-04a, b and c)
288 contain zoned microcrysts (which are very close to 500 μm size) with a core enriched in both Fo and
289 NiO (e.g. $Fe_{86.6-85.4}$ and 0.36-0.33 wt % NiO). These values are similar to those from forsterite cores in
290 porphyritic basalt varieties. Two macrocrysts with repetitive zonation contain Fe-enriched core and rim
291 with similar composition ($Fe_{79.0-80.2}$, 0.07-0.10 wt % NiO) and mantle which is slightly enriched in Mg
292 ($Fe_{82.1-82.9}$, 0.18-0.23 wt % NiO). Angular to sub-angular forsterite fragments are generally rare, but
293 abundant in one sample (A19-08; Fig. 7h). The composition of these fragments is highly variable ($Fe_{86.3-}$
294 75.4 ; 0.32-0.10 wt % NiO). Besides, minor Ca, Mn and traces of Cr were detected (Table 2) with generally
295 high Ca and low Mn in high-Fo olivine and low Ca and high Mn in low-Fo olivine.

296

297

298 **Clinopyroxene**

299 Clinopyroxene occurs as euhedral to subhedral macro-, microcrysts and groundmass crystals, with a
300 single megacryst (Fig. 9, Supplementary Data Fig. S2). Compared to forsterite, clinopyroxene occurs
301 more often as fragments of broken crystals. Most clinopyroxene grains are colourless, with few crystals
302 containing greenish cores. In some samples microcrysts define a trachytoid texture and rarely they occur
303 as glomerocrysts. Based on variable shapes, zonation patterns, internal structures, and compositions, we
304 distinguish six populations of clinopyroxene. In total 342 spot analyses were obtained from 12 samples.
305 Most analyses indicate diopsidic to augitic compositions, with rarely hedenbergite and aegirine-augite
306 compositions in greenish cores.

307 Population I (common) is represented by cores of euhedral to subhedral macro- and microcrysts
308 (Fig. 9a, b, d), typically not in optical continuity with the mantle or rim zones. These cores are
309 characterized by a variable morphology (round, oval, embayed) and sometimes show strong resorption
310 at their boundary with the mantle or rim. Their Mg# values (calculated as $100 \cdot \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$) and
311 contents of Cr, Al and Ti point to three varieties (Ia, Ib and Ic - Table 3). Variety Ia is augite ($\text{Di}_{60-68}\text{Hed}_{12-16}\text{others}_{19-25}$) with Mg# between 85 and 80, low Cr (always below detection limit) and high Al
312 (3.2-3.8 wt % Al_2O_3) and Ti (1.3-1.7 wt % TiO_2). Variety Ib ($\text{Di}_{51-63}\text{Hed}_{14-20}\text{others}_{22-33}$) has Mg# of 81-
313 73, contains higher Cr (0.05-0.4 wt % Cr_2O_3) and high, but variable Al and Ti (2.3-4.7 wt % Al_2O_3 and
314 0.6-1.6 wt % TiO_2). Variety Ic ($\text{Di}_{68-76}\text{Hed}_{6-11}\text{others}_{16-21}$) with high Mg# values of 94-87 is rather Cr-rich
315 (0.4-1.4 wt % Cr_2O_3) but depleted in Al and Ti (0.3-0.5 wt % TiO_2 and 0.8-1.8 wt % Al_2O_3). Varieties
316 Ia and Ic contain similar amounts of Na_2O (0.5-0.6 and 0.4-0.7 wt % respectively), variety Ib is slightly
317 enriched in Na (typically 0.7-1.0 wt % Na_2O , with two spot analysis of 0.5 wt %).

319 Population II (less common, but abundant in some samples) occurs as rounded and embayed
320 cores in macrocrysts (Fig. 9c). A prominent feature is their large size, typically 1-4 mm, and up to 9 x 8
321 mm and their irregular extinction. Unlike population I, these crystals do not have a mantle zone. Augite
322 ($\text{Di}_{60-75}\text{Hed}_{8-14}\text{others}_{16-29}$) has a relatively high Mg# (90-83) and contains elevated Cr (0.3-0.7 wt %
323 Cr_2O_3). Other minor components are Al (1.3-2.9 wt % Al_2O_3), Ti (0.4-1.0 wt % TiO_2) and Na (0.5-0.7
324 wt % Na_2O) (Table 3).

325 Population III (common) is augite ($\text{Di}_{54-74}\text{Hed}_{10-21}\text{others}_{13-30}$) with a characteristic spongy texture

326 (Fig. 9d, e) that occurs as cores in euhedral to subhedral macro- and microcrystals or as mantle around
327 augite I. In the latter case it forms either complete or discontinuous mantle zone. The mineral has a
328 variable Mg# (86-74), contains relatively little Cr (0.1-0.4 wt % Cr₂O₃ with one spot analysis of 0.8 wt
329 %) and variable amounts of Al (1.6-4.0 wt % Al₂O₃), Ti (0.5-1.9 wt % TiO₂) and Na (0.3-0.9 wt %
330 Na₂O) (Table 3).

331 Despite the observed compositional differences between augite I, II and III, there is a common
332 feature for these three populations – a negative correlation between values of Mg# and content of Al
333 and Ti (Figs. 11-12). The behavior of Cr is different: augite with Mg# between 85 and 73 contains Cr at
334 approximately similar and relatively low level, while high Mg# augite (94-85) shows a positive
335 correlation between Mg# and Cr (Fig. 10).

336 Population IV (generally rare, but common in some samples) occurs as greenish cores in
337 euhedral to subhedral micro- and rarely macrocrystals, typically not in optical continuity with the rim of
338 the crystals (Fig. 9f, g). Some of these cores contain inclusions of Ti-rich garnet (schorlomite), titanite
339 and/or apatite (Figs. 9f, 18b). Clinopyroxene IV is compositionally very diverse (Di₁₈₋₇₆Hed₁₀₋₆₂Aeg₃₋
340 _{20others₂₋₂₃}); minerals are characterized by a highly variable Mg# values (86-26) and Al, Ti and Na
341 contents (0.4-5.1 wt % Al₂O₃, 0.3-2.2 wt % TiO₂ and 0.8-3.8 wt % Na₂O) (Table 3). The Cr contents are
342 below the detection limit with the exception of one crystal (Fig. 9g) that contains 0.17-0.24 wt % Cr₂O₃.
343 Also, this crystal contains a mantle zone while others do not.

344 Population V (common) occurs as two varieties. One occurs as mantles around cores of
345 populations I, III and rarely IV (Fig. 9a, b, d, e, g) and generally shows oscillatory zoning, with rare
346 continuous growth zonation. Another represents euhedral to subhedral macro- (rare), microcrystals
347 (common) and groundmass (common) crystals with oscillatory-type or rarely sector-type zoning (Fig.
348 9h, i). This augite (Di₅₆₋₇₆Hed₆₋₂₁others₁₂₋₃₀) has high Mg# values with a range from 93 to 73 and the high
349 Cr contents (up to 1.4 wt % Cr₂O₃), which decrease towards the rims (Table 3, (Fig. 10). In contrast, Al
350 and Ti increase from the inner mantle/core (1.2 wt % Al₂O₃ and 0.4 wt % TiO₂) towards the outer
351 mantle/rim (5.7 wt % Al₂O₃ and 3.6 wt % TiO₂). These compositional variations are related to the size
352 of the crystals: macrocrystals have a core with higher Mg# values and lower Al-Ti contents compared
353 with microcrystals and groundmass crystals, which have lower Mg# and higher Al-Ti contents. Variable

354 Na (0.2-0.7 wt % Na₂O) correlate positively with Cr (Fig. 13), suggesting a minor kosmochlor
355 NaCr(Si₂O₆) component (up to 4 mol.%). However, molar Na/Cr ratio is always ≥ 1 (up to 9.2),
356 suggesting that parts of Na is incorporated as aegirine and/or jadeite components (Fig. 13).

357 Population VI (common) is augite (Di₄₉₋₆₉Hed₁₂₋₂₆others₁₄₋₂₉), which forms thin rims around
358 almost all macro-, microcrysts and groundmass crystals (Fig. 9). The latter (particularly those < 50 μm)
359 can be completely composed by this population. This augite has variable Mg# values (85-69), contains
360 low Cr (commonly below detection limit), and is characterized by high and variable contents of Al (1.7-
361 6.1 wt % Al₂O₃), (1.3-3.6 wt % TiO₂) (Table 3, Figs. 10-12) and variable Na (0.3-0.7 wt % Na₂O) with
362 no obvious relationship between Na and Cr (Fig. 13).

363

364 **Spinel**

365 Opaque minerals compose an essential part of the groundmass. They are rarely present as microcrysts
366 and also occur as inclusions in forsterite and augite macrocrysts. The crystals are different in
367 morphology, size, internal zonation, and compositional data show the presence of two distinct solid
368 solutions series (magnesiocromite-chromite_{ss} and magnetite-ulvöspinel_{ss}) (Table 4, Figs. 14, 15).

369 Euhedral spinel inclusions (up to 180 x 90 μm in size) in forsterite macrocrysts with normal
370 zonation and microcrysts (up to 160 x 110 μm in size) are considered to be the earliest opaque phases
371 to crystallize (Fig. 14a). They are Ti-poor (1.7-3.3 wt % TiO₂) and Al-rich (6.3-8.9 wt % Al₂O₃)
372 magnesiocromite-chromite_{ss}. From core to rim contents of Mg and Cr decrease, while Al and Ti
373 increase (Fig. 15). Spinel inclusions (up to 30 μm in size) in forsterite with reverse zonation are
374 characterized by high and variable Ti (12.1-21.7 wt % TiO₂) and low Cr (3.5-8.5 wt % Cr₂O₃) contents
375 (Fig. 15), while having similar Mg and Al (5.4-7.8 wt % MgO and 7.9-9.6 wt % Al₂O₃) as euhedral
376 chromite (see above).

377 Strongly zoned euhedral to subhedral spinel-group minerals occur as rare microcrysts (up to 230
378 x 100 μm) and groundmass crystals (Fig. 14b, c). They consist of a chromite core rimmed by
379 “titanomagnetite” (magnetite-ulvöspinel_{ss}). Chromite is a low Ti variety (2.0-3.8 wt % TiO₂) with
380 elevated Al (6.5-10.4 wt % Al₂O₃) and Mg (3.6-9.6 wt % MgO). The boundary between chromite and
381 magnetite-ulvöspinel_{ss} is gradational with considerable compositional changes over a distance of 5-10

382 μm (Fig. 15). Magnetite-ulvöspinel_{ss} is Ti-rich (11.2-26.7 wt % TiO_2), Cr-poor (1.5-15.7 wt % Cr_2O_3)
383 and contains less Al (1.1-4.6 wt % Al_2O_3) and Mg (2.0-4.0 wt % MgO) compared to the core chromite
384 (Fig. 15). One of the studied samples (A19-64) contains texturally and compositionally similar zoned
385 microcrysts with chromite cores and magnetite-ulvöspinel_{ss} rims, but the latter contains ilmenite
386 exsolution lamellae (Fig. 14d). All magnetite-ulvöspinel_{ss} in this sample, either occurring as rim on
387 chromite or as euhedral to subhedral crystals in groundmass contain exsolved ilmenite lamellae. Cr-
388 bearing magnetite-ulvöspinel_{ss} (2.2-10.4 wt % Cr_2O_3 , 14.2-19.9 wt % TiO_2 , Fig. 15) occurs as inclusions
389 in clinopyroxene (typically in varieties II, III, V and VI). Groundmass spinel with grain sizes between
390 1 and 50 μm (rarely up to 100 μm) occurs as euhedral to subhedral crystals (Fig. 14e). Compositionally
391 they are magnetite-ulvöspinel_{ss} with low Cr (0.2-3.8 wt % Cr_2O_3), and high Ti (18.2 to 25.6 wt % TiO_2 ;
392 Fig. 15). Few crystal cores are relatively enriched in Cr (4.1-9.9 and 16.1-17.9 wt % Cr_2O_3).

393 All studied samples contain additional anhedral and rather homogeneous spinel (up to 330 x 260
394 μm ; Fig. 14f) with Mg, Al and Ti (2.3-4.1 wt % MgO , 1.7-4.0 wt % Al_2O_3 and 14.5-21.8 wt % TiO_2) at
395 levels similar to groundmass magnetite-ulvöspinel (Fig. 15). However, these grains are very low in Cr
396 (mostly below the detection limit, Fig. 15) and compositionally different grains are present within a
397 single specimen, e.g. the sample A19-08 contains crystals (150 – 300 μm in size) with quite different Ti
398 content (14.5-14.7, 17.0-17.1 and 20.0-20.8 wt % TiO_2).

399

400 **Plagioclase**

401 Plagioclase occurs as subhedral tabular laths to anhedral grains (mostly 20 to 70 μm in size, and rarely
402 up to 200 μm in length; Fig. 16a) that form polycrystalline aggregates (Supplementary Data Fig. S3).
403 Plagioclase is andesine-oligoclase solid solution ($\text{An}_{0.28-0.37}\text{Ab}_{0.58-0.66}\text{Or}_{0.04-0.06}$) with predominance of
404 andesine compositions (Table 5); they contain minor Sr (0.4-0.6 wt % SrO) and Ba (0.2-0.4 wt % BaO).
405 A single microcryst contains a strongly corroded core with labradorite composition ($\text{An}_{0.66}\text{Ab}_{0.33}\text{Or}_{0.01}$),
406 rimmed by Ca-enriched andesine ($\text{An}_{0.42}\text{Ab}_{0.55}\text{Or}_{0.03}$). It is unclear if this core is a xenocryst or a primary
407 crystallized phase, but with subsequent reaction with a melt. In addition to microcrysts, sample A19-08
408 contains relatively large embayed plagioclase grains (250 x 180 to 520 x 490 μm ; Fig. 16b) of oligoclase
409 composition ($\text{An}_{0.20-0.29}\text{Ab}_{0.67-0.71}\text{Or}_{0.04-0.09}$).

410 **Apatite**

411 Apatite is a common accessory mineral in all studied samples and occurs as needle- or spindle-like
412 crystals up to 120 μm in length and 10 μm in thickness (Fig. 16c, Supplementary Data Fig. S4). Because
413 of small crystals size WDS analysis was not possible, and SEM/EDS analysis show that the mineral is
414 close to an ideal fluorapatite with minor Si (ca. 1.0 wt % SiO_2). No peaks from other cations have been
415 found on ED spectra.

416

417 **Other minerals**

418 A sample from the Oldoinyo Naisiusiu hill (A19-08) contains rare, but relatively large grains of
419 perovskite (up to 160 x 210 μm in size) (Fig. 18a), a mineral which is not a typical phase in basaltic
420 rocks. It is always surrounded by a 10-50 μm thick rim of Mg-bearing ilmenite. The latter, in turn,
421 contains numerous tiny inclusions of REE-rich mineral(s) (not precisely identified). In addition, the
422 sample contains grains of green clinopyroxene with inclusions of schorlomite, titanite and apatite (Fig.
423 18b). These were also observed as rare minerals in some of the other samples.

424

425 **Glass**

426 Glass is an essential component in the studied samples (Supplementary Data Fig. S3). Fresh, unaltered
427 glass is, however, rare (Fig. 16d), but well preserved in two samples from Oldoinyo Emusenge hill
428 (A16-05 and A16-06). The glass contains 58.7-60.2 wt % SiO_2 , 18.9-19.4 wt % Al_2O_3 and 13.3-14.6 wt
429 % $\text{Na}_2\text{O}+\text{K}_2\text{O}$ (Table 5) and corresponds to a phonolite on TAS classification diagram (Fig. 4a). It is
430 strongly depleted in Mg (0.5-0.7 wt % MgO) and Ca (0.4-0.9 wt % CaO) and slightly enriched in Fe
431 (4.4-4.7 wt % $\text{Fe}_2\text{O}_{3\text{total}}$). SEM/EDS analysis shows presence of P (up to 0.5 wt % P_2O_5), Cl (up to 0.4
432 wt %) and S (up to 0.2 wt % SO_3). The analytical totals are typically between 95 and 98 wt % (Table 5)
433 and this can be an indication of presence of other volatile components, e.g. H_2O . During alteration the
434 glass was transformed to a fine-grained assemblage of montmorillonite, K-Na feldspar, $\text{Na}\pm\text{K}$ zeolite
435 and sodalite (Fig. 17).

436

437

438 Sr-Nd ISOTOPES

439

440 Limited published Sr and Nd isotope compositions for the Ogol basalts are controversial. While the data
441 presented by Mollel (2007) are relatively uniform ($^{87}\text{Sr}/^{86}\text{Sr} = 0.70371\text{-}0.70381$ and $^{143}\text{Nd}/^{144}\text{Nd} =$
442 $0.51255\text{-}0.51261$), the data from Adelsberger *et al.* (2011) show wide variations ($^{87}\text{Sr}/^{86}\text{Sr} = 0.70350\text{-}$
443 0.70577 and $^{143}\text{Nd}/^{144}\text{Nd} = 0.51221\text{-}0.51257$). The highest Sr (0.70577) and lowest Nd (0.51221) ratios
444 were found in sample OL-22E (Adelsberger *et al.*, 2011), which is compositionally quite different from
445 all other Ogol basalts (“microbasalt” with low Mg, Cr and Ni contents and high Ti and P contents, see
446 above) and is probably not directly related with them. In the present study we analysed 10 samples of
447 the Ogol basalts for their Sr and Nd isotopic composition; complemented by three lava samples from
448 the Lemagarut volcano and three samples of basement samples (Table 6). All measured $^{87}\text{Sr}/^{86}\text{Sr}$ and
449 $^{143}\text{Nd}/^{144}\text{Nd}$ values were corrected for an age of 2.3 Ma.

450 Our new data (Fig. 19) confirm variable isotopic composition of the studied basalts ($^{87}\text{Sr}/^{86}\text{Sr}_{\text{initial}}$
451 $= 0.70377\text{-}0.70470$ and $^{143}\text{Nd}/^{144}\text{Nd}_{\text{initial}} = 0.51246\text{-}0.51261$). The data show a negative correlation
452 between $^{87}\text{Sr}/^{86}\text{Sr}_{\text{initial}}$ and $^{143}\text{Nd}/^{144}\text{Nd}_{\text{initial}}$ ratios, with a sample from Oldoinyo Emusenge hill (A16-05)
453 showing lowest $^{87}\text{Sr}/^{86}\text{Sr}_{\text{initial}}$ and highest $^{143}\text{Nd}/^{144}\text{Nd}_{\text{initial}}$ and a sample from Oldoinyo Mati hill (A19-
454 02) haven the highest $^{87}\text{Sr}/^{86}\text{Sr}_{\text{initial}}$ and lowest $^{143}\text{Nd}/^{144}\text{Nd}_{\text{initial}}$ values.

455 Sr-Nd isotopic data for basalts to trachyandesites from the contemporaneously erupting
456 Lemagarut volcano partly overlap with those for the Ogol basalts, forming a continuous field (Table 6;
457 Fig. 19). One sample of Lemagarut trachyandesite (Mollel, 2007), has a low $^{143}\text{Nd}/^{144}\text{Nd}$ ratio of
458 0.51219, which is quite different from all other samples. The reason for low $^{143}\text{Nd}/^{144}\text{Nd}$ ratio is
459 unknown, but can potentially be explained by alteration. Significant changes in Nd isotope compositions
460 were observed for nephelinitic rocks from the nearby Sadiman volcano, where highly altered
461 nephelinitic tuffs have much low $^{143}\text{Nd}/^{144}\text{Nd}$ ratio compared with fresh, unaltered tuffs and lavas
462 (Zaitsev *et al.*, 2019).

463 In contrast to Ogol and Lemagarut, basaltic rocks from the Ngorongoro volcano are
464 characterized by irregular distribution of measured $^{87}\text{Sr}/^{86}\text{Sr}$ and $^{143}\text{Nd}/^{144}\text{Nd}$ ratios (Fig. 19). Majority
465 of samples from Ngorongoro show similar, to Ogol and Lamagarut, ranges for Sr and Nd isotopic

466 compositions ($^{87}\text{Sr}/^{86}\text{Sr} = 0.70405\text{-}0.70533$ and $^{143}\text{Nd}/^{144}\text{Nd} = 0.51227\text{-}0.51277$), and three samples have
467 extremely high $^{87}\text{Sr}/^{86}\text{Sr}$ ratio of 0.70634, 0.70801 and 0.71317 (Nonnotte, 2007; Mollel *et al.*, 2008).

468 Importantly, the Sr and Nd isotopic data for the Ogol basalts, as well as the basaltic rocks from
469 Lamagarut and Ngorongoro correlate with their major element whole-rock data (Fig. 20). The $^{87}\text{Sr}/^{86}\text{Sr}$
470 ratios correlate positively with SiO_2 , Al_2O_3 , Na_2O , K_2O and negatively with MgO , TiO_2 , Fe_2O_3 and CaO ,
471 while $^{143}\text{Nd}/^{144}\text{Nd}$ ratio shows the opposite relationships.

472

473 **DISCUSSION**

474

475 Basaltic magmatism is a common phenomenon for the Eastern branch of the East African Rift and
476 numerous publications present detailed discussions on the origin and evolution of basaltic magmas based
477 on whole-rock geochemical (major and trace elements) and radiogenic isotope (Sr, Nd and Pb) data (see
478 recent reviewers by Rooney 2017; 2020a; b; c; d). However, only few publications (e.g. Paslick *et al.*,
479 1996) provide detailed textural and mineralogical descriptions of the basaltic rocks, which is, however,
480 extremely important for sound genetic interpretations of multi-reservoir melting and contamination
481 processes, as is evident from the observations presented above.

482

483 **Distinguishing different crystal populations**

484 The textural and compositional differences between the various populations of olivine, clinopyroxene,
485 spinel group minerals and plagioclase in the studied samples (see above) show that besides primary
486 minerals (autocrysts and microlites) which crystallized from a particular melt batch erupted as lava,
487 variable amounts of their crystal load is xenocrystic.

488 As primary minerals we consider groundmass phases and some, but not all, macro- and
489 microcrysts:

490 (1) Forsterite macro- and microcrysts with normal zonation and unzoned microcrysts (Fig. 7a, b).

491 (2) Clinopyroxene cores and mantles with characteristic oscillatory-type, rarely sector-type or
492 continuous growth zonation (population V) and augite rims (population VI; Fig. 9).

493 (3) Magnesiochromite-chromite as inclusions in forsterite with normal zonation and microcrysts,

494 magnetite-ulvöspinel rims around those and discrete magnetite-ulvöspinel crystals in groundmass (Fig.
495 14a-e).

496 (4) Microcrysts and groundmass andesine-oligoclase and apatite (Fig. 16a,c).

497 All other crystals were probably entrained from additional sources and are considered to
498 represent xenocrystic material, however, some of them can represent antecrysts:

499 (1) Forsterite crystals (and angular fragments thereof) that show reverse and repetitive zonations (Fig.
500 7c-f) and their Cr-bearing magnetite-ulvöspinel inclusions.

501 (2) Rounded and embayed clinopyroxene cores (populations I-IV) and angular fragments thereof (Fig.
502 9a-g), including perovskite, rimmed by ilmenite and schorlomite ± apatite and titanite (Figs. 9f, 18a-b)
503 that occur as inclusions in population IV (green cores).

504 (3) Anhedral, low-Cr magnetite-ulvöspinel (Fig. 14f) and embayed oligoclase macro- and microcrysts
505 (Fig. 16b).

506 The presence of abundant and texturally/compositionally diverse xenocrystic minerals suggest
507 an open-system and multi-stage crystallization history for the Ogol basalts and the involvement of
508 several sources during their evolution and these are considered in the following.

509

510 **Potential sources of xenocrystic minerals**

511 Importantly, macrocrystic forsterite (normal zonation) and xenocrystic forsterite (reverse zonation)
512 occurs in the same sample, and rim compositions for both types are very similar (Fig. 8). This indicates
513 that xenocrystic forsterite cores were entrained in the magma before eruption with a residence time long
514 enough for partial equilibration. Unzoned angular fragments (occasionally with partial rims only) point
515 to entrainment close to the surface, only shortly before lava eruption and with no time for equilibration
516 with the surrounding melt. A forsterite megacryst with Fe-rich core and normal zonation (see above)
517 was also likely entrained before eruption, because the crystal rim has a quite different composition from
518 rims on normally zoned Fe-poor forsterite in the same sample. Forsterite with repetitive zonation (Fe-
519 rich core – Fe-depleted mantle – Fe-rich rim, Fig. 7e, f), although being extremely rare, is example of
520 multi-stage growth in melt (or melts) with variable Mg#.

521 All studied forsterite crystals show elevated contents of Ca and Mn (0.17-0.42 wt % CaO and

522 0.12-0.38 wt % MnO) compared to mantle forsterite ($Fo_{\geq 90}$) which has typically up to 0.1 wt % and
523 rarely up to 0.2 wt % oxides (e.g. Dawson, 2002; Koornneef *et al.*, 2009). Therefore, we assume that
524 the Ogol lavas contain basaltic forsterite only and none of the xenocrysts derive from mantle lithologies.
525 Two possible scenarios can explain the occurrence of both low and high Fe forsterite: (1) mixing of
526 relatively primitive magma with evolved magma batches related to the differentiation of the primary
527 Ogol magmas and (2) mixing with evolved melts derived from the Lemagarut volcanic system, which
528 was active at the same time as the Ogol magmas erupted. The second scenario is supported by the
529 occurrence of embayed oligoclase crystals in the Ogol basalts (Fig. 16b), as their morphology and
530 composition suggest that they represent xenocrysts. Lavas of basaltic trachyandesite with plagioclase as
531 phenocrysts have been observed in the Laetoli area (samples A16-21 and A16-28, Supplementary Data
532 Table S1) and on the western lower slope of the Lemagarut volcano; our unpublished data show that
533 these plagioclase phenocrysts are compositionally similar to those found in the Ogol basalts.

534 Green core clinopyroxenes (population IV) with inclusions of schorlomite, titanite and apatite
535 (Figs. 9f, 18b), as well as discrete perovskite crystals (Fig. 18a) provide strong evidence for the
536 additional contribution of evolved alkaline sources - either by mixing between the basaltic Ogol melts
537 and nephelinitic melts or by assimilation/fragmentation of already crystallized alkaline rocks (foidolitic)
538 at depth. However, no nephelinitic rocks with an age of ca. 2.3 Ma are known in the Laetoli area. Here,
539 alkaline magmatism (Sadiman volcano, Laetoli tuffs, Engolsin cone) is older (>3 Ma) and we can only
540 speculate that alkaline magmas were still present in mantle after 3.0 Ma and mixed with Ogol melts.
541 This assumption is somewhat supported by alkaline volcanoes in the Gregory Rift, which were active at
542 2.3 Ma, e.g. Tarosero volcano (Mana *et al.*, 2015; Braunger *et al.*, 2020, submitted). At Crater Highlands
543 alkaline magmatism was reactivated around 1.2 Ma (Fig. 1) with the Embagai volcano being active at
544 1.2-0.8 Ma (Fig. 2). Alternatively, the minerals can simply derive from small-scale assimilation of
545 unexposed rocks related to the Sadiman volcano or even from the Laetoli tuffs by rising basaltic melts.
546 One of the studied green core crystals (Fig. 9g) is different from the others as it shows a distinct patchy
547 zonation, contains 0.17-0.24 wt % Cr_2O_3 , and is mantled by augite population V. Note that all other
548 green cores are rimmed by augite VI only and some grains even do not have the rim. This suggests that
549 entrainment of this crystal happened at a different depth level and time, but the detailed origin of this

550 crystal remains unclear. Augite population Ia (Fig. 9a) could be also related to an alkaline source because
551 of the lack of Cr. It is always mantled by augite V, these crystals had a long residence time in the melt
552 and therefore, another, probably deeper mixing event, compared to clinopyroxene IV, is considered.

553 Rounded and embayed clinopyroxene cores (populations Ib, Ic) are mantled by primary augite
554 V followed by a rim of augite VI, while clinopyroxene II cores are rimmed by augite VI but lack a
555 mantle of augite V. This may suggest different residence times in melt(s). Compositionally, they are
556 also different, but plotting all together they show gradational changes. From augite Ic, via augite II, to
557 augite Ib we observe decreasing Mg# (94-73) and Cr₂O₃ (from 1.4 to 0.05 wt %), accompanied by
558 increasing Al₂O₃ (from 0.8 to 4.7 wt %) and TiO₂ (from 0.3 to 1.6 wt %) content (Figs. 10-12). These
559 variations suggest minerals crystallization from melts with different compositions. All cores of augite
560 Ib and Ic, and even megacrystic augite II are relatively homogeneous (Fig. 9) with small compositional
561 variations within a single crystal. We suggest that they crystallized in closed-system intermediate
562 magma chambers under equilibrium conditions. The compositional variations in augite Ib, Ic and II are
563 similar with those of primary augite V (Figs. 10-12). This indicates basaltic melt compositions that may
564 derive from the Lemagarut magmatic system, and were mixed with Ogol melts.

565 Alternatively, high Mg# and Cr augite Ic may derive from mantle rocks. Although mantle
566 xenoliths are not known from basaltic or nephelinitic rocks of the Crater Highlands, they occur, for
567 example, at Lashaine volcano within the Gregory Rift, at Labait volcano located on the eastern edge of
568 the Tanzanian Craton and other localities. Clinopyroxene from Lashaine (Dawson, 2002) is enriched in
569 Cr and Na (1.7-3.1 wt % Cr₂O₃ and 1.2-2.5 wt % Na₂O) compared with mineral from Ogol, but
570 clinopyroxene occurring in glassy melt pockets, veinlets and veins shows similar levels of Cr and Na
571 (0.2-1.8 wt % Cr₂O₃ and 0.3-0.8 wt % Na₂O) as augite Ic. However, as no “true” mantle minerals (e.g.
572 low Ca-Mn and high Fo olivine, high Cr-Na clinopyroxene and/or orthopyroxene) were found in the
573 Ogol basalts we suggest a basaltic source is more likely for augite Ic.

574 Spongy clinopyroxene cores and mantles (population III) are compositionally similar to augite
575 populations Ib, II and even partly overlap with augite Ic (Fig. 9d, e) and are always overgrown by augite
576 V, indicating relatively long residence time in the melts. The large compositional variations for augite
577 III (Figs. 10-12) are interpreted to represent magmatic differentiation. Spongy- or sieve-textured

578 minerals, mainly plagioclase and pyroxene, are known in various volcanic rocks and two models were
579 suggested to explain origin of the texture (e.g. Nelson & Montana, 1992; Stewart & Pearce, 2004; Shane
580 *et al.*, 2019). The formation of spongy-textured crystals can be related to magma mixing processes or
581 due to rapid decompression during melt ascent.

582

583 **Primary minerals**

584 Primary orthomagmatic phenocrysts of forsterite and clinopyroxene (population V) are Mg-rich (Fo_{89.5}
585 and Mg# of 94, respectively) and crystallized from rather primitive melts with Mg# of 72.5-70.5 during
586 forsterite crystallization (assuming crystal/melt equilibrium with $K_{d_{Fe/Mg}(ol/liq)} = 0.30-0.35$, depending
587 of pressure; Ulmer, 1989). Primary forsterite with lower Mg contents (down to Fo_{84.2}) reflect magmatic
588 differentiation, magma mixing processes, or combinations thereof.

589 Similarly, the compositional variation of augite V is explained, characterized by well-developed
590 zonation with a decrease of Mg# and Cr from core to rim (in case of mantle from inner to outer parts)
591 and a simultaneous increase in Al and Ti (Fig. 11, 12). Augite VI that occurs as rims around pheno- and
592 microphenocrysts and as groundmass crystals, formed somewhat later but from the same evolving
593 magma. Based on its textural context and evolved composition (low Mg# and high content of Al and
594 Ti), we suggest that augite VI crystallized immediately before and during eruptions of melts.

595 Two compositionally distinct spinel group minerals crystallized during magmatic
596 differentiation. The change from magnesiochromite-chromite_{ss} to magnetite-ulvöspinel_{ss} indicates not
597 only changes in melt composition, but is further related to changes in physical-chemical conditions, such
598 as temperature and oxygen fugacity (Hill & Roeder, 1974).

599

600 **Ogol primary melt composition**

601 The relatively high concentrations of Cr and Ni in the studied samples allow them to be considered as
602 candidates for primary magma compositions, not only for the Ogol rocks, but also for the Lemagarut
603 and even Ngorongoro volcanoes (see above). The Ogol basalts are relatively primitive (Table 1), but
604 show continuous variations in contents of major and trace elements (Fig. 4). With 11.5 wt % MgO (Mg#
605 = 61), 975 ppm Cr and 222 ppm Ni, a sample from Oldoinyo Emusenge hill (A16-05; Fig. 6a) could

606 approximate the composition of a primary magma in equilibrium with its mantle source. Out of all
607 studied samples, A16-05 contains the most Mg-rich olivine ($\text{Fo}_{89.5-87.5}$) and clinopyroxene V (Mg# of
608 94-85) and magnesiochromite. Also, it is characterized by the lowest $^{87}\text{Sr}/^{86}\text{Sr}_i$ (0.703768) and highest
609 $^{143}\text{Nd}/^{144}\text{Nd}_i$ (0.512614) ratios compared to the Ogot as well as Lemagarut basalts. However, even this
610 sample contains few high-Fe forsterite as well as xenocrystic augite (populations Ib, Ic, II and IV).
611 Therefore, we suggest that its composition is close but not exactly corresponds to a primary melt
612 composition. The presence of relatively Fe-rich and Ni-poor forsterite causes a decrease of Mg and Ni
613 and an increase of Fe compared to an initial composition (Fig. 21). Likewise, abundant augite xenocrysts
614 mainly influence Mg, Fe and Cr content of the whole rock. The lack of trace element data for primary
615 and xenocrystic minerals, do not allow a more detailed discussion on possible changes, particularly for
616 trace elements, of melts initial composition.

617 Despite the modification of the studied samples by variable amounts of xenocrystic minerals
618 (see above), their minor and trace elements distribution is similar to that of OIB (Fig. 5), except for Ba
619 and Cs. The relative depletion of the Ogot basalts in K could reflect partial melting in the presence of a
620 K-bearing phase (amphibole and/or phlogopite) as suggested for other volcanic rocks in the NTD
621 zone (e.g. Mana *et al.*, 2015). Both minerals are known in mantle xenoliths from Eledoi and Pello
622 Hill where they form metasomatic veins (Dawson & Smith, 1988) and giant mantle-derived
623 amphibole and mica megacrysts (up to 12 cm in size) occur at Deeti cone (Johnson *et al.*, 1977;
624 Zaitsev *et al.*, 2013). The high values of $((\text{Tb}/\text{Yb})_{\text{CN}} = 2.48-2.83)$ observed in the Ogot basalts indicate
625 melting of enriched lithosphere in the presence of residual garnet (Wang *et al.*, 2002). Such a mantle
626 source for the basalts is also supported by relatively high values of Ce/Pb ratio of 17.3-24.9.

627

628 **Crystallization depth and temperature of the Ogot basalts**

629 Based on the clinopyroxene-only barometer of Nimis & Ulmer (1998), there is no clear evidence for a
630 deep (mantle) origin of the various xenocrysts (populations I-IV). Also, pressure estimates for
631 xenocrysts and orthomagmatic clinopyroxene (populations V and VI) are rather similar, with many of
632 them illustrating negative, and thus, geologically meaningless crystallization depths though
633 (Supplementary Data Fig. S5). Apparently, such pressure estimates are not suitable in this case. It is

634 known that melt composition influences clinopyroxene composition as well. Therefore, we used augite
635 populations V and VI to constrain the crystallization pressure and temperature by applying the
636 clinopyroxene-liquid barometer of Neave & Putirka (2017) combined with the thermometer equation 33
637 from Putirka (2008).

638 As a proxy for the liquid composition in equilibrium with augite, we used whole-rock
639 compositions, being aware that (i) the whole rock composition cannot represent the coexisting liquid for
640 both augite populations V and VI, as these crystallized over certain intervals and partly during different
641 stages, and that (ii) mineral removal as well as accumulation/xenocryst entrainment result in
642 discrepancies between equilibrium melts and the whole rock compositions. Correction of the whole rock
643 composition by removing xenocrystals was rejected because the amount of xenocrysts cannot be
644 determined exactly. We still applied the thermobarometer, since the required liquid components of the
645 pressure equation solely involve the Al, Na, K and Si concentration (Neave & Putirka, 2017). The
646 aluminum and alkali metal budget of the whole rock compositions is mainly controlled by plagioclase,
647 and the relatively high SiO₂ concentration of the basalts (~45 wt %) acts as a buffer, hence diminishing
648 the effect of olivine, clinopyroxene or spinel contamination. Note that the temperature equation 33
649 (Putirka, 2008) depends on further liquid components (Ca, Mg, Fe, Ti concentration), on which
650 xenocryst contamination, however, has little effect (Supplementary Data Fig. S6). For obtaining reliable
651 results, two important criteria for the pressure calculations have to be satisfied (e.g. Putirka *et al.*, 1996;
652 Putirka, 2008; Neave & Putirka, 2017). Firstly, the Fe and Mg concentrations of clinopyroxene and the
653 coexisting liquid must be in accordance with the assumed Fe-Mg exchange coefficient of $K_D[\text{Fe-Mg}]^{\text{cpx-}}$
654 $\text{liq}=0.27\pm 0.3$ (Fig. 22a). Equally important is that the observed augite composition conforms with the
655 predicted values (Fig. 22b), for which we used 5 percentage points as cutoff. If these requirements are
656 not fulfilled, clearly meaningless pressure variations down to -25 kbar are calculated (Fig. 22c). Data
657 that meet these requirements (Fig. 22c), however, suggest formation depths for augite V typically
658 between 1 and 4 kbar (uncertainty of ± 1.4 kbar) at a temperature of ~1150-1220 °C (uncertainty of \pm
659 45 °C). The calculated pressures indicate upper-crustal crystallization depths for augite V of
660 approximately 3-12 km, which is consistent with several basaltic magmas at other localities (e.g.
661 Eimeishan; Icelandic rift zone; Tao *et al.*, 2015; Neave & Putirka, 2017). Groundmass augite VI does

662 not meet the equilibrium constraints, which is related to the relatively evolved composition of late stage
663 augite VI compared to early magmatic augite V (Fig. 22a). The groundmass assemblage of augite VI,
664 forsterite and magnetite-ulvöspinel_{ss}, however, would be suitable for calculating equilibrium
665 temperature, silica activity and oxygen fugacity using the QUIF software package (Andersen *et al.*,
666 1993). While late stage augite and magnetite-ulvöspinel_{ss} are in textural equilibrium, there is, however,
667 no clear evidence that forsterite crystallized contemporaneously. Indeed, the high uncertainty (>100 °C)
668 of the calculated temperature and oxygen fugacity (above 3 log units), as well as the failure to calculate
669 the silica activity (iteration limit) clearly points towards chemical disequilibrium between forsterite on
670 the one hand, and augite VI + magnetite-ulvöspinel_{ss} on the other hand (Table 7). Hence, we consider
671 the calculated results as not reliable.

672

673 **Magma mixing and contamination of the Ogot basalt**

674 The variability of the Ogot basalts in mineralogy, whole rock compositions, incompatible trace element
675 ratios (e.g. Ce/Pb, La/Sm) and Sr-Nd isotopic ratios reflect the contribution from different components,
676 two of them (basaltic and alkaline) were suggested above. Furthermore, assimilation of crustal rocks
677 composing the Tanzania Craton may contribute as considered for other volcanic rocks in the Gregory
678 rift and Crater Highlands (e.g. Mana *et al.*, 2015; Mollel *et al.*, 2008).

679

680 Archean basement exposed in the southwestern part of the NTD zone close to Laetoli, allows to
681 consider granite-gneisses and associated garnet amphibolites (Table 1S) as possible contaminants.
682 Garnet amphibolite is unlikely to explain the observed $^{87}\text{Sr}/^{86}\text{Sr}$ and $^{143}\text{Nd}/^{144}\text{Nd}$ variations as up to 25
683 % assimilation would be needed to cover this range and such a voluminous addition would be reflected
684 in the whole rock composition of the Ogot basalts (Fig. 19). Assimilation of granite-gneisses would only
685 require <3 % bulk assimilation but the assimilation line between Ogot basalts and granite-gneiss is
686 shifted to higher $^{143}\text{Nd}/^{144}\text{Nd}$ values. An alternative explanation for the observed variations is the
687 addition of an alkaline source, for which data from the Sadiman volcano were used (sample Sad-10;
688 Zaitsev *et al.*, 2012). A plot of $(\text{La}/\text{Sm})_{\text{N}}$ against $^{87}\text{Sr}/^{86}\text{Sr}$ and $^{143}\text{Nd}/^{144}\text{Nd}$ ratios with data from
689 assimilation and fractional crystallization (AFC) modeling between Ogot basalt, Tanzanian Craton

690 granite-gneiss and Sadiman nephelinite suggests involvement up to 5% of alkaline component (Fig. 23).
691 The combination of contamination with granite-gneisses and mixing with alkaline melts or
692 contamination by unexposed Sadiman plutonic rocks can explain the observed geochemical
693 characteristics. Involvement of alkaline source is further supported by relatively high alkalinity ratios
694 (molar Na+K)/Al) of 0.98-1.07 of phonolitic glass and xenocrystic green core pyroxene (population IV)
695 with perovskite and schorlomite (see above). At the same time, abundant xenocrysts (high-Fe forsterite,
696 diverse clinopyroxene cores and mantles, embayed plagioclase) suggest that mixing with other basaltic
697 melts is a further important process in evolution of Ogol melts.

698 Much of the compositional variation of the Ogol basalts (Fig. 4) could be explained by crystal
699 fractionation processes and subsequent mixing between different melt batches which related to
700 fractionation of primary Ogol magma only. However, as shown above, whole rock data show that major
701 and trace elements correlate with variable Sr and Nd isotope ratios (Figs. 19, 20). This is strong evidence
702 against closed-system evolution of Ogol melts, because Sr and Nd isotopic ratios should not change
703 during differentiation under closed-system condition. Although, we cannot completely exclude that
704 fractionation of primary Ogol magmas took place, our mineralogical and geochemical data suggest that
705 mixing with melts related to the Lemagarut volcanic system is more likely process. Rocks composing
706 the Lemagarut volcano vary from relatively primitive basaltic rocks (Mg# = 58-56 and 537-495 ppm
707 Cr) to highly evolved trachyandesite (Mg# = 27-30 and 15-3.4 ppm Cr) (Mollet, 2007). Our data show
708 that the former contains olivine with compositions similar to Ogol phenocrysts with reverse zonation
709 and the latter is rich in oligoclase-andesine_{ss} phenocrysts. Diagrams showing relationships between
710 major elements and isotopic ratios (Figs. 19, 20) support mixing between Ogol and Lemagarut derived
711 melts. For example, mixing with evolved Lemagarut basaltic trachyandesite (Supplementary Data Table
712 S1) would explain the enrichment of Zr and Hf observed for some Ogol basalts (Fig. 5a).

713 The detailed relationships between Ogol and Lemagarut may be more complex. Although Hay
714 (1987) assumed that Ogol lavas erupted from a series of individual vents, Mollet *et al.* (2011) provide
715 some arguments in favor of close genetical relation of Ogol and Lemagarut eruptions and suggested that
716 Ogol basalts may represent distal flows from the Lemagarut volcano. Limited mineralogical data for the
717 Lemagarut rocks (Mollet *et al.*, 2011) does not allow to discuss possible genetic relations with Ogol

718 rocks. At this stage we can only speculate that the volcanic series of Ogol and Lemagarut may record
719 an uniform evolutionary history reflected by the composition of a single parental melt and the onset of
720 cotectic crystallization of olivine, pyroxene, magnetite and plagioclase. Linear correlations in major
721 element abundances in Ogol and Lemagarut lavas are unlikely to reflect crystallization along a liquid
722 line of descent. Simple AFC modelling indicates that Lemagarut derivatives could be result of
723 fractionation of 50 – 60 % Ogol melt with addition of 5 % of the crustal material. A detailed study of
724 the Lemagarut volcano is needed to prove or disprove this idea.

725

726 CONCLUSIONS

727

728 The textural, mineralogical and geochemical heterogeneity of the Ogol basalts suggests that these
729 represent hybride rocks that formed under open-system conditions. The major findings we consider to
730 be important for formation and evolution of the Ogol basalts (Fig. 24) include

731 (1) formation of a primary Ogol melts during partial melting of amphibole- (and phlogopite?)
732 and garnet-bearing lithospheric material;

733 (2) although small degrees of fractional crystallization may explain the compositional variations
734 of the studied samples, the presence of abundant xenocrysts (Fe-rich olivine, clinopyroxene with
735 variable composition, plagioclase, perovskite and schorlomite) and variable Sr and Nd initial isotopic
736 ratios suggest that magma mixing with other melts from different sources was important;

737 (3) there is no clear evidence for a deep (mantle) origin of the various clinopyroxene xenocrysts.
738 Early orthomagmatic augite (population V) crystallized at a temperature of ~1150-1220 °C, and an
739 upper-crustal depth of 3-12 km (1-4 kbar). Late stage augite (population VI) and magnetite-ulvöspinel_{ss}
740 are in textural equilibrium, while thermodynamic calculations indicate that groundmass forsterite did
741 not form contemporaneously;

742 (4) magma mixing between Ogol and Lemagarut melts can explain the observed chemical
743 variations of the Ogol basalts. Magma mixing was not a single-stage process but happened at different
744 depths with variable crystals residence time in melts during ascent to the surface;

745 (5) mixing with mantle-derived alkaline melts and/or contamination with alkaline rocks

746 crystallized beneath the Sadiman volcano or even from the Laetolil tuffs did occur, but was a relatively
747 minor process;

748 (6) small-scale contamination of the Ogol melts by crustal granite-gneisses may have occurred.
749 However, this assumption is based on geochemical modeling only, and no mineralogical evidence for
750 this was found.

751 Overall, the Ogol basalts, particularly from Oldoinyo Emusenge volcanic cone, seem to be the
752 best examples of near-primary basaltic melts within the Crater Highlands region. They could also be
753 considered as parental to the Lemgarut volcano, but more data are needed to solve this question.

754

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762

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770

771 **SUPPLEMENTARY DATA**

772

773 Supplementary data are available at Journal of Petrology online.

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Table 1: Whole-rock geochemical data for the studied Ogol basalts

Sample	A16-05	A16-06	A16-25	A16-26	A19-01	A19-02	A19-06	A19-07	A19-08	A19-64	A19-4c	A19-4b
Latitude	741220	740683	739808	745223	748761	748497	740867	742655	734090	741124	747201	747201
Longitude	9642114	9642261	9644377	9643513	9641516	9641340	9639339	9641864	9643749	9635170	9640489	9640489
SiO ₂ , wt %	45.90	45.16	46.19	45.60	46.68	44.95	45.03	46.31	46.68	47.27	46.61	47.12
TiO ₂	3.06	3.54	2.99	2.98	2.80	2.77	3.37	3.01	2.75	2.70	2.85	2.86
Al ₂ O ₃	7.59	8.45	10.00	8.99	9.44	9.61	8.94	9.96	9.85	10.13	10.16	10.00
Fe ₂ O _{3total}	14.37	15.45	13.72	14.07	13.79	13.47	14.53	13.90	13.17	13.44	13.47	13.23
MnO	0.18	0.19	0.18	0.19	0.17	0.18	0.19	0.18	0.18	0.18	0.21	0.18
MgO	11.50	10.97	8.54	9.58	10.22	10.46	9.42	9.60	9.36	8.82	9.15	9.24
CaO	12.32	11.23	10.82	12.13	11.30	9.98	12.04	10.81	11.03	10.35	10.98	11.04
Na ₂ O	2.27	1.90	2.81	2.19	2.16	2.97	2.09	2.76	2.67	2.62	2.80	2.89
K ₂ O	1.06	1.00	1.37	1.25	1.11	1.20	1.16	1.35	1.30	1.19	1.44	1.37
P ₂ O ₅	0.34	0.41	0.43	0.39	0.39	0.39	0.40	0.41	0.45	0.38	0.49	0.50
S _{total}	0.89	0.03	0.09	0.05	0.02	0.02	0.01	0.02	0.02	0.01	0.02	0.02
LOI	0.89	1.36	2.36	2.14	1.48	3.66	2.25	1.22	2.09	2.53	1.29	1.02
Total	99.48	99.69	99.50	99.56	99.54	99.66	99.43	99.53	99.55	99.62	99.45	99.47
Mg#	61	58	55	57	59	61	56	58	58	57	57	58
Li, ppm	7.60	7.21	6.98	8.21	7.19	7.35	9.13	8.34	9.05	9.55	11.32	9.57
Be	1.43	1.38	1.91	1.51	1.51	1.72	1.45	2.09	1.81	1.58	2.09	1.91
Sc	36.3	25.7	26.7	27.0	29.5	30.1	28.9	28.0	27.8	29.1	29.3	30.3
V	293	266	205	232	257	251	280	245	228	214	238	245
Cr	975	519	497	542	479	450	520	450	458	520	572	569
Co	78.4	57.4	53.7	55.9	70.0	60.3	64.9	61.1	58.9	62.9	63.2	64.2
Ni	222	197	171	183	183	165	190	174	172	185	196	198
Cu	190	143	138	139	147	166	169	143	152	138	175	189
Zn	118	105	114	106	110	109	114	112	107	118	111	120
Ga	15.2	13.0	17.7	14.7	16.6	13.5	18.3	15.0	14.8	16.1	15.2	16.1
Rb	20.1	15.8	22.8	19.5	21.2	20.6	20.3	23.7	13.7	17.5	25.3	22.6
Sr	560	608	846	757	614	924	665	765	733	855	827	842
Y	19.1	19.2	21.6	19.5	19.7	18.6	19.7	20.7	20.2	20.2	22.2	22.8
Zr	194	186	362	358	179	192	198	208	203	202	218	233
Nb	50.5	47.8	59.2	52.5	53.8	52.8	54.6	60.4	62.8	58.0	66.1	69.1
Cs	0.15	0.18	1.42	0.13	0.22	0.13	0.14	0.16	0.31	0.13	0.20	0.27
Ba	781	362	910	854	622	658	987	1076	732	638	979	1130
La	38.2	42.7	52.1	47.5	47.3	47.2	45.8	52.3	54.1	46.9	59.2	60.9

Ce	86.9	84.6	99.9	89.6	88.7	93.5	94.1	104	105	93.5	119	119
Pr	10.2	10.9	12.1	11.1	11.2	10.7	11.0	11.7	11.8	10.7	13.1	13.1
Nd	41.8	44.6	48.6	44.5	41.9	42.6	44.0	46.5	45.9	41.1	51.6	50.8
Sm	8.43	9.24	9.51	9.01	8.19	8.25	8.84	8.87	8.77	8.05	9.52	9.67
Eu	2.50	2.53	2.86	2.61	2.32	2.43	2.57	2.68	2.65	2.42	2.90	2.88
Gd	7.04	7.48	7.90	7.25	6.88	6.82	7.11	7.28	7.10	6.70	7.81	7.73
Tb	0.96	0.98	1.02	0.95	0.98	0.90	1.02	1.03	0.97	0.95	1.07	1.07
Dy	4.74	4.93	5.38	4.82	4.69	4.70	4.95	4.91	4.96	4.64	5.17	5.27
Ho	0.88	0.81	0.86	0.79	0.85	0.90	0.90	0.94	0.96	0.88	1.06	1.00
Er	1.97	2.15	2.36	2.11	2.01	2.00	2.09	2.18	2.07	2.07	2.24	2.21
Tm	0.27	0.25	0.30	0.25	0.27	0.29	0.29	0.31	0.30	0.30	0.31	0.32
Yb	1.54	1.59	1.87	1.58	1.67	1.63	1.66	1.76	1.70	1.72	1.82	1.75
Lu	0.20	0.21	0.25	0.21	0.20	0.21	0.21	0.23	0.23	0.22	0.25	0.24
Hf	5.29	5.42	9.41	9.30	5.51	4.96	5.56	5.58	5.29	5.03	5.49	5.58
Ta	2.99	3.25	3.55	3.25	2.96	3.24	3.24	3.63	3.60	3.26	3.80	3.72
Pb	5.03	4.82	4.01	5.07	4.81	5.00	4.71	5.49	5.48	4.69	6.59	5.77
Th	5.08	5.27	6.61	5.89	5.50	6.21	5.64	6.76	7.18	6.01	7.69	7.48
U	0.89	1.05	1.46	0.85	1.12	1.27	0.85	0.97	1.38	1.23	0.84	0.89

LOI - loss on ignition. Mg# = $100 * \text{Mg} / (\text{Mg} + 0.9 * \text{Fe}_{\text{total}})$. Map datum WGS84, Latitude and longitude in UTM coordinate system, zone 36M.

Table 2: Selected compositions of forsterite

Sample Zonation Crystal Analysis	16-05 normal			16-06			19-04c			16-05 reverse			16-06			19-07			16-06 unzoned			19-08																		
	pheno	core	26	rim	9	3	pheno	core	5	rim	20	micro	core	151	rim	16	pheno	core	19	rim	19	pheno	core	71	rim	73	micro	core	52	gm	core	13	fragment	core	139	fragment	core	145		
	41.60	41.13	40.02	40.61	39.64	40.65	40.10	41.42	40.53	38.23	39.68	38.38	39.53	39.72	40.26	39.65	39.65	40.83	39.41																					
SiO ₂	41.60	41.13	40.02	40.61	39.64	40.65	40.10	41.42	40.53	38.23	39.68	38.38	39.53	39.72	40.26	39.65	40.83	39.41																						
TiO ₂			0.03				0.03	0.03	0.03																															
Al ₂ O ₃		0.03	0.05				0.03																																	
FeO	9.25	9.93	15.95	10.99	17.38	13.22	15.26	12.95	17.26	24.65	16.29	23.42	17.66	19.93	17.28	15.69	16.19	21.49																						
MgO	48.62	48.23	43.05	48.37	43.42	45.73	44.03	45.72	42.35	37.50	44.00	38.26	43.02	40.13	42.38	44.63	43.32	38.66																						
MnO	0.16	0.15	0.30	0.13	0.25	0.19	0.23	0.20	0.33	0.30	0.21	0.26	0.23	0.32	0.32	0.21	0.18	0.29																						
CaO	0.31	0.34	0.42	0.34	0.36	0.29	0.25	0.33	0.31	0.27	0.37	0.27	0.34	0.17	0.34	0.31	0.26	0.25																						
NiO	0.33	0.28	0.17	0.33	0.20	0.23	0.22	0.35	0.25	0.05	0.21	0.14	0.20	0.13	0.21	0.25	0.20	0.11																						
Cr ₂ O ₃	0.08	0.10	0.05	0.09	0.11	0.06	0.03	0.06	0.06																															
Total	100.34	100.18	100.03	100.86	101.36	100.37	100.16	101.05	101.14	101.00	100.76	100.79	100.98	100.40	100.87	100.80	101.00	100.21																						
Si	1.013	1.007	1.009	0.994	0.995	1.008	1.007	1.018	1.016	0.996	0.996	0.996	0.996	1.015	1.012	0.993	0.997	1.017																						
Ti			0.001				0.001	0.001	0.001																															
Al		0.001	0.001				0.001	0.001	0.001																															
Total	1.013	1.008	1.011	0.994	0.995	1.008	1.008	1.019	1.018	0.996	0.996	0.998	0.996	1.015	1.013	0.993	0.997	1.017																						
Mg	1.765	1.761	1.619	1.766	1.624	1.691	1.648	1.675	1.582	1.456	1.647	1.480	1.617	1.529	1.589	1.666	1.624	1.487																						
Fe	0.188	0.203	0.336	0.225	0.365	0.274	0.320	0.266	0.362	0.537	0.342	0.508	0.372	0.426	0.363	0.329	0.362	0.464																						
Mn	0.003	0.003	0.006	0.003	0.005	0.004	0.005	0.004	0.007	0.007	0.004	0.006	0.005	0.007	0.007	0.004	0.005	0.006																						
Ca	0.008	0.009	0.011	0.009	0.010	0.008	0.007	0.009	0.008	0.008	0.010	0.008	0.009	0.005	0.009	0.008	0.010	0.007																						
Ni	0.006	0.006	0.003	0.006	0.004	0.005	0.005	0.007	0.005	0.001	0.004	0.003	0.004	0.003	0.004	0.005	0.004	0.004																						
Cr	0.002	0.002	0.001	0.002	0.002	0.001	0.001	0.001	0.001	2.008	2.008	2.005	2.007	1.969	1.973	2.014	2.006	1.966																						
Total	1.973	1.984	1.977	2.010	2.010	1.983	1.986	1.962	1.965	2.008	2.008	2.005	2.007	1.969	1.973	2.014	2.006	1.966																						
Fo	89.5	88.8	81.9	87.8	80.8	85.3	83.0	85.4	80.5	72.5	82.0	73.8	80.5	77.6	80.5	82.7	81.0	75.6																						

Fo – forsterite end-member, mol. %.

Table 3: Selected analyses of clinopyroxene

Sample Population Analysis	16-05	16-06	16-25	16-26	19-04a	16-06	19-04c	19-07	19-08	19-64	16-26	19-07	19-08	19-04a	19-08	16-06		
	1a 26	31	1b 43	33	1c 1	49	113	II 97	123	3	102	53	110	IV 14	6	21	23	
SiO ₂	50.19	49.32	51.33	53.41	53.89	53.44	54.62	53.68	54.32	52.90	50.99	53.18	52.88	51.58	47.61	51.69	50.74	50.76
TiO ₂	1.67	1.57	1.30	0.63	0.51	0.42	0.42	0.64	0.50	0.84	1.93	0.83	0.93	0.52	1.99	0.51	0.95	0.97
Al ₂ O ₃	3.34	3.82	4.00	2.35	1.81	1.05	1.32	2.50	1.85	2.56	3.23	2.30	2.14	1.26	5.10	0.69	3.36	3.40
Cr ₂ O ₃			0.21	0.31	1.37	0.89	0.53	0.38	0.71	0.32	0.20	0.23	0.32				0.24	0.17
Fe ₂ O ₃	3.03	4.36			2.29	2.29				0.41	0.16	0.78	0.78	3.33	4.44	6.32	4.06	3.54
FeO	5.31	4.36	8.65	6.68	3.54	2.22	4.01	5.57	4.63	5.72	7.77	6.53	4.52	10.93	8.15	14.07	6.46	4.50
MnO	0.17	0.21	0.18	0.13	0.09	0.09	0.10	0.11	0.10	0.13	0.18	0.12	0.11	0.45	0.31	0.80	0.29	0.17
MgO	14.20	13.97	13.69	16.16	17.14	18.06	18.28	17.08	17.42	15.88	13.41	15.60	16.18	8.42	8.75	4.46	12.52	15.36
NiO				0.03	0.09		0.05	0.04	0.05		0.08	0.04	0.03					
CaO	21.88	22.06	19.63	19.24	21.39	21.43	20.74	19.96	20.75	20.95	20.86	22.37	22.41	21.25	22.47	17.35	20.34	20.32
Na ₂ O	0.58	0.57	0.81	0.53	0.70	0.50	0.50	0.65	0.62	0.65	0.48	0.46	0.41	1.84	1.27	3.71	1.38	0.75
Total	100.37	100.25	99.80	99.47	100.52	100.39	100.57	100.61	100.95	100.36	100.76	100.26	100.73	99.58	100.08	99.60	100.34	99.94
Si	1.859	1.832	1.908	1.972	1.952	1.940	1.971	1.947	1.960	1.934	1.966	1.953	1.927	1.977	1.812	2.017	1.889	1.874
Al	0.141	0.167	0.092	0.028	0.048	0.045	0.029	0.053	0.040	0.066	0.034	0.113	0.047	0.023	0.188		0.110	0.126
Fe ³⁺						0.015												
Total	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.017	2.000	2.000
Mg	0.784	0.774	0.759	0.890	0.926	0.977	0.984	0.923	0.937	0.866	0.740	0.854	0.879	0.481	0.496	0.259	0.695	0.846
Fe ²⁺	0.164	0.136	0.269	0.206	0.107	0.067	0.121	0.169	0.140	0.175	0.240	0.200	0.138	0.350	0.259	0.459	0.201	0.139
Fe ³⁺	0.085	0.121				0.048				0.011	0.007	0.004	0.021	0.096	0.127	0.186	0.114	0.098
Al	0.005		0.083	0.074	0.029		0.027	0.054	0.039	0.045	0.022	0.027	0.053	0.019	0.034	0.041	0.032	0.022
Ti	0.047	0.044	0.036	0.017	0.014	0.011	0.011	0.017	0.013	0.023	0.012	0.054	0.026	0.015	0.057	0.015	0.027	0.027
Mn	0.005	0.007	0.006	0.004	0.003	0.003	0.003	0.003	0.003	0.004	0.003	0.004	0.004	0.014	0.010	0.027	0.009	0.005
Cr			0.006	0.009	0.039	0.026	0.015	0.011	0.020	0.009	0.014	0.006	0.007	0.003	0.010	0.027	0.009	0.005
Ni				0.001	0.003		0.002	0.001	0.001		0.002	0.001	0.001				0.007	0.005
Ca	0.868	0.878	0.782	0.761	0.830	0.833	0.802	0.776	0.802	0.821	0.811	0.887	0.875	0.873	0.916	0.725	0.811	0.804
Na	0.042	0.041	0.058	0.038	0.049	0.035	0.035	0.045	0.044	0.046	0.035	0.032	0.029	0.137	0.093	0.281	0.100	0.054
Total	2.000	2.000	2.000	2.000	1.999	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	1.983	2.000	2.000
Mg#	83	85	74	81	90	94	89	85	87	83	90	81	86	58	66	36	78	86

Table 3: Continued

Sample Population Analysis	16-05 V (mantle)		19-07		16-26		16-05 V (core)		19-06		16-26		16-05 VI		19-06		16-06		19-07		16-26	
	28	27	25	91	46	44	60	61	62	30	32	73	74	3	31	39	104	78				
SiO ₂	52.24	51.10	52.41	54.39	52.43	53.38	52.20	50.27	52.63	50.04	51.86	52.30	54.08	53.37	48.94	50.29	49.04	46.44				
TiO ₂	0.76	1.31	0.91	0.55	1.26	0.53	1.12	1.30	0.74	1.79	1.19	1.30	0.81	0.97	1.99	1.87	2.59	3.65				
Al ₂ O ₃	1.91	2.20	1.24	1.76	2.45	1.58	1.67	2.97	1.46	2.94	2.31	1.65	1.71	1.81	2.91	3.01	4.40	4.87				
Cr ₂ O ₃	1.33	0.57	0.46	1.06	0.62	1.09	0.52	1.24	0.61	0.45	0.98	0.22	0.56	0.36	0.06	0.06	0.05	0.05				
Fe ₂ O ₃	1.37	2.90	2.18	0.29	0.29	0.29	0.52	3.30	2.16	3.13	0.38	0.79	0.63	2.42	4.06	1.48	1.40	3.30				
FeO	3.86	3.45	3.62	4.41	5.23	5.23	6.24	3.25	4.40	4.14	4.83	5.79	6.53	7.03	5.50	6.17	6.85	8.69				
MnO	0.09	0.10	0.10	0.09	0.12	0.17	0.12	0.10	0.14	0.12	0.11	0.12	0.17	0.14	0.16	0.11	0.16	0.18				
MgO	16.56	15.71	16.31	17.39	15.57	16.45	14.87	15.73	17.58	14.76	15.43	15.82	15.66	15.11	13.98	14.07	12.93	11.46				
NiO				0.03									0.03									
CaO	21.14	22.46	22.87	21.11	22.51	19.96	22.01	21.12	20.04	22.54	22.24	21.85	20.69	21.24	21.65	22.03	22.09	20.57				
Na ₂ O	0.56	0.42	0.30	0.65	0.41	0.60	0.37	0.61	0.44	0.43	0.46	0.35	0.47	0.44	0.43	0.48	0.56	0.68				
Total	99.82	100.22	100.40	101.43	100.88	98.99	99.12	99.89	100.21	100.34	99.79	100.19	100.71	100.48	99.62	99.57	100.02	99.90				
Si	1.919	1.882	1.922	1.955	1.914	1.975	1.947	1.857	1.925	1.851	1.914	1.926	1.981	1.965	1.838	1.877	1.831	1.764				
Al	0.081	0.096	0.054	0.045	0.086	0.025	0.053	0.129	0.063	0.128	0.086	0.072	0.019	0.035	0.129	0.123	0.169	0.218				
Fe ³⁺	0.022	0.024	0.024	0.024	0.024	0.024	0.024	0.014	0.012	0.020	0.002	0.002	0.002	0.002	0.034	0.002	0.002	0.018				
Total	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000				
Mg	0.907	0.863	0.892	0.932	0.847	0.907	0.827	0.866	0.959	0.814	0.849	0.869	0.855	0.829	0.783	0.783	0.720	0.649				
Fe ²⁺	0.119	0.106	0.111	0.132	0.160	0.162	0.195	0.100	0.135	0.128	0.149	0.178	0.200	0.216	0.173	0.193	0.214	0.276				
Fe ³⁺	0.038	0.058	0.036	0.008	0.008	0.008	0.008	0.078	0.048	0.067	0.011	0.020	0.055	0.043	0.081	0.041	0.039	0.076				
Al	0.002	0.002	0.002	0.029	0.020	0.044	0.021	0.036	0.020	0.050	0.014	0.036	0.022	0.027	0.056	0.009	0.025	0.104				
Ti	0.021	0.036	0.025	0.015	0.035	0.015	0.031	0.036	0.004	0.004	0.033	0.004	0.005	0.004	0.005	0.052	0.073	0.104				
Mn	0.003	0.003	0.003	0.003	0.004	0.005	0.004	0.003	0.004	0.004	0.003	0.004	0.005	0.004	0.005	0.004	0.005	0.006				
Cr	0.039	0.017	0.013	0.030	0.018	0.032	0.015	0.036	0.018	0.013	0.029	0.006	0.016	0.011	0.002	0.002	0.002	0.002				
Ni				0.001									0.001									
Ca	0.832	0.887	0.899	0.813	0.881	0.792	0.880	0.836	0.785	0.893	0.879	0.862	0.812	0.838	0.871	0.881	0.884	0.837				
Na	0.040	0.030	0.021	0.045	0.029	0.043	0.027	0.044	0.031	0.031	0.033	0.025	0.033	0.031	0.031	0.035	0.041	0.050				
Total	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000				
Mg#	88	89	89	88	84	85	81	90	88	86	85	83	81	79	82	80	77	70				

Fe₂O₃ and FeO calculated from charge balance (4 cations and 6 O), Mg# = 100 * Mg / (Mg + Fe²⁺).

Table 4: Selected analyses of spinel-group minerals

Sample Variety	16-05		19-06		19-07		19-06		19-04c		16-05		19-06		19-04a		19-08		19-04a	
	core	rim	core	rim	core	rim	core	rim	core	rim	core	rim	core	rim	core	rim	core	rim	core	rim
Analysis	1	4	30	32	62	8	12	93	96	39	40	73	49	50	44	46	9	88	23	23
MgO	12.69	8.51	9.82	9.89	5.44	9.58	3.07	4.81	3.53	5.49	2.07	3.94	2.51	1.69	2.55	1.84	2.28	2.58	3.24	3.24
Al ₂ O ₃	7.65	7.17	8.38	8.40	7.89	7.03	4.46	7.45	1.80	8.28	3.44	2.81	6.10	2.69	5.02	3.01	1.23	1.99	3.05	3.05
SiO ₂	0.09	0.09	0.07	0.08	0.09	0.06	0.05	0.05	0.03	0.07	0.08	0.03	0.03	0.10	0.04	0.08	0.04	0.12	0.07	0.07
CaO	0.12	0.12					0.16	0.05	0.08		0.12	0.08	0.04	0.05	0.06	0.14				
TiO ₂	1.77	2.12	3.19	3.19	15.30	2.06	19.41	2.73	18.01	3.47	23.39	14.19	15.14	25.57	18.48	25.00	22.06	17.13	21.17	21.17
V ₂ O ₅	0.10	0.09	0.21	0.18	0.57	0.12	0.35	0.12	0.26	0.20	0.33	0.28	0.32	0.35	0.36	0.37	0.32	0.60	0.38	0.38
Cr ₂ O ₃	53.91	50.81	44.62	44.04	6.17	51.56	9.61	44.17	3.27	42.48	2.02	10.36	17.88	0.75	9.91	1.06	1.79	0.04	0.08	0.08
MnO	0.11	0.19	0.16	0.14	0.32	0.14	0.56	0.50	0.65	0.25	0.63	0.55	0.49	0.70	0.56	0.65	0.74	0.56	0.68	0.68
NiO	0.18	0.13	0.15	0.18	0.08	0.14	0.03	0.05	0.09	0.12	0.04	0.09	0.03	0.03	0.07	0.07	0.07	0.05	0.05	0.05
ZnO	0.02	0.10	0.07	0.06	0.07	0.10	0.08	0.24	0.07	0.10	0.09	0.13	0.15	0.09	0.10	0.08	0.10	0.11	0.08	0.08
Fe ₂ O ₃	7.95	9.20	12.54	13.11	26.66	8.93	17.90	12.64	30.27	12.12	17.71	29.10	14.82	14.99	17.78	15.88	24.08	34.18	26.45	26.45
FeO	14.99	21.55	20.76	20.65	38.30	19.80	44.59	27.38	41.84	27.65	49.20	37.82	41.69	51.49	44.47	50.93	47.49	42.87	45.92	45.92
Total	99.47	100.07	99.96	99.90	100.88	99.50	100.28	100.15	99.90	100.21	99.11	99.38	99.18	98.49	99.34	99.03	100.22	100.22	101.16	101.16
Fe ²⁺	0.367	0.557	0.499	0.496	0.706	0.506	0.812	0.726	0.782	0.705	0.863	0.761	0.845	0.881	0.842	0.874	0.847	0.837	0.803	0.803
Mg	0.625	0.430	0.490	0.494	0.280	0.484	0.164	0.250	0.192	0.282	0.113	0.213	0.135	0.093	0.137	0.101	0.125	0.141	0.173	0.173
Mn	0.003	0.005	0.005	0.004	0.009	0.004	0.017	0.015	0.020	0.007	0.019	0.017	0.015	0.022	0.017	0.020	0.023	0.017	0.021	0.021
Ni	0.005	0.003	0.004	0.005	0.002	0.004	0.001	0.001	0.003	0.003	0.001	0.003	0.001	0.001	0.000	0.000	0.002	0.002	0.002	0.002
Zn	0.001	0.003	0.002	0.001	0.002	0.002	0.002	0.006	0.002	0.002	0.002	0.003	0.004	0.003	0.003	0.002	0.003	0.003	0.002	0.002
Total	1.000	0.999	1.000	1.000	1.000	1.000	0.996	0.998	0.998	1.000	0.998	0.998	1.000	1.000	0.999	0.997	1.000	1.000	1.000	1.000
Cr	1.408	1.362	1.182	1.167	0.168	1.380	0.272	1.216	0.094	1.159	0.058	0.298	0.509	0.022	0.283	0.031	0.052	0.001	0.002	0.002
Fe ³⁺	0.198	0.235	0.316	0.331	0.693	0.228	0.482	0.331	0.831	0.315	0.488	0.796	0.401	0.418	0.484	0.439	0.665	0.942	0.713	0.713
Al	0.298	0.286	0.331	0.332	0.321	0.280	0.188	0.306	0.078	0.336	0.149	0.120	0.259	0.117	0.214	0.130	0.053	0.086	0.129	0.129
Ti	0.044	0.054	0.080	0.080	0.398	0.052	0.522	0.072	0.494	0.090	0.645	0.388	0.410	0.713	0.503	0.691	0.609	0.472	0.570	0.570
Fe ²⁺	0.047	0.054	0.083	0.083	0.401	0.054	0.522	0.072	0.494	0.093	0.645	0.388	0.410	0.715	0.503	0.691	0.610	0.477	0.572	0.572
V	0.003	0.002	0.006	0.005	0.016	0.003	0.010	0.003	0.008	0.006	0.010	0.008	0.009	0.010	0.011	0.011	0.010	0.018	0.011	0.011
Total	1.997	1.994	1.998	1.998	1.997	1.998	1.996	2.000	1.998	1.998	1.994	1.998	1.998	1.995	1.997	1.994	1.998	1.996	1.997	1.997
Si	0.003	0.003	0.002	0.003	0.003	0.002	0.002	0.002	0.001	0.002	0.003	0.001	0.001	0.004	0.001	0.003	0.001	0.004	0.002	0.002
Ca		0.005					0.006	0.002	0.003		0.005	0.003	0.001	0.002	0.002	0.006				

Varieties (1) – microphenocrystals, inclusions in forsterite (Fo = 84-89 mol.%), (2) – inclusions in forsterite (Fo = 74-78 mol.%), (3) zoned microphenocrystals and inclusions in forsterite (Fo = 84-87 mol.%), (4) – inclusions in dropside, (5) groundmass crystals, (6) xenocrystals. Fe₂O₃ and FeO calculated from charge balance (3 cations and 4 O).

Table 5: Selected analyses of plagioclase and glass

Sample Analysis	16-05 plagioclase					16-26					16-05 glass				
	2	6	12	1	3	5	1	3	5	1	11	15			
SiO ₂	60.01	59.90	61.23	59.49	60.69	61.81	57.56	58.96	58.72	57.56	58.96	58.72			
TiO ₂							0.47	1.11	0.85	0.47	1.11	0.85			
Al ₂ O ₃	24.81	24.56	23.53	25.31	24.27	24.00	19.06	18.54	18.76	19.06	18.54	18.76			
Fe ₂ O ₃	0.56	0.80	0.77	0.57	0.51	0.57	4.29	4.58	4.30	4.29	4.58	4.30			
CaO	7.25	6.87	6.01	7.72	6.53	5.76	0.87	0.44	0.84	0.87	0.44	0.84			
MgO	0.04	0.05	0.04	0.06	0.03	0.02	0.71	0.49	0.52	0.71	0.49	0.52			
MnO		0.02			0.02	0.02	0.08	0.09	0.07	0.08	0.09	0.07			
Na ₂ O	6.55	6.97	7.47	6.76	7.17	7.52	8.75	7.47	7.55	8.75	7.47	7.55			
K ₂ O	0.74	0.83	0.97	0.70	0.90	1.12	5.52	5.53	5.47	5.52	5.53	5.47			
SrO	0.51	0.65	0.45	0.50	0.49	0.45									
BaO	0.19	0.23	0.23	0.27	0.26	0.45									
P ₂ O ₅							0.03	0.03	0.05	0.03	0.03	0.05			
SO ₃							0.49	0.51	0.40	0.49	0.51	0.40			
Cl															
-O=Cl ₂							0.27	0.28	0.35	0.27	0.28	0.35			
Total	100.66	100.87	100.71	101.38	100.87	101.71	98.03	97.96	97.96	98.03	97.96	97.96			
Si	2.674	2.672	2.727	2.642	2.700	2.727									
Al	1.303	1.291	1.235	1.325	1.273	1.248									
Fe ³⁺	0.019	0.027	0.026	0.019	0.017	0.019									
Total	3.996	3.990	3.988	3.986	3.989	3.993									
Ca	0.346	0.328	0.287	0.367	0.311	0.272									
Na	0.566	0.603	0.645	0.582	0.618	0.643									
K	0.042	0.047	0.055	0.040	0.051	0.063									
Mn	0.000	0.001	0.000	0.000	0.001	0.001									
Mg	0.003	0.003	0.003	0.004	0.002	0.002									
Sr	0.013	0.017	0.012	0.013	0.013	0.011									
Ba	0.003	0.004	0.004	0.005	0.005	0.008									
Total	0.973	1.003	1.006	1.011	1.001	1.000									

P, S and Cl were not analysed in plagioclase.

Table 6: Rb-Sr and Sm-Nd data for samples of the Ogol and Lemagarut lavas and the Precambrian rocks

Sample	Rb ppm	Sr ppm	$^{87}\text{Rb}/^{86}\text{Sr}_m$	$^{87}\text{Sr}/^{86}\text{Sr}_m$	$^{87}\text{Sr}/^{86}\text{Sr}_f$	Sm ppm	Nd ppm	$^{147}\text{Sm}/^{144}\text{Nd}_m$	$^{143}\text{Nd}/^{144}\text{Nd}_m$	$^{143}\text{Nd}/^{144}\text{Nd}_i$	ϵ_{Nd}
Ogol lavas											
A16-05	20.9	562	0.1077	0.703771±9	0.703768	8.17	40.8	0.1211	0.512616±6	0.512614	-0.41
A16-06	17.1	735	0.0670	0.703903±15	0.703901	9.79	51.2	0.1158	0.512579±3	0.512577	-1.13
A16-25	24.2	786	0.0890	0.703907±17	0.703904	9.69	52.5	0.1117	0.512553±4	0.512551	-1.63
A16-26	23.3	845	0.0796	0.704088±10	0.704085	9.28	49.8	0.1126	0.512561±3	0.512559	-1.48
A19-02	23.9	991	0.0696	0.704704±11	0.704702	8.59	43.6	0.1191	0.512462±4	0.512460	-3.41
A19-4c	28.2	851	0.0958	0.704367±10	0.704364	9.98	54.6	0.1106	0.512511±5	0.512509	-2.45
A19-06	24.3	752	0.0931	0.704180±16	0.704177	9.39	48.7	0.1166	0.512572±3	0.512570	-1.26
A19-07	27.0	823	0.0948	0.704075±9	0.704072	9.46	50.9	0.1124	0.512546±3	0.512544	-1.77
A19-08	15.2	784	0.0561	0.704325±9	0.704323	9.10	49.5	0.1111	0.512493±5	0.512491	-2.80
A19-64	19.4	881	0.0635	0.704359±9	0.704357	8.33	43.7	0.1152	0.512541±5	0.512539	-1.87
Lemagarut lavas											
A16-21	56.2	1043	0.1557	0.705000±10	0.704995	14.4	84.7	0.1029	0.512422±3	0.512420	-4.19
A16-28	69.6	1086	0.1852	0.704949±13	0.704943	13.1	76.0	0.1039	0.512421±3	0.512419	-4.21
A16-29	46.1	1397	0.0953	0.704980±19	0.704977	11.7	63.3	0.1117	0.512414±2	0.512412	-4.34
Precambrian basement											
A16-10	4.08	124	0.0950	0.704687±11	0.704684	4.00	13.9	0.1741	0.512378±5	0.512375	-5.07
A16-12	222	146	4.4560	0.862926±14	0.862783	3.80	21.8	0.1053	0.511058±5	0.511056	-30.79
A16-16c	17.0	336	0.1465	0.709289±14	0.709284	3.27	15.5	0.1276	0.511452±6	0.511450	-23.12

A16-10 – green schist, A16-12 – granite-gneiss, A16-16c – garnet amphibolite. M – measured, I – initial (calculated for 2.3 Ma). $^{143}\text{Nd}/^{144}\text{Nd}$ ratio is normalized to $^{146}\text{Nd}/^{144}\text{Nd} = 0.7219$, error is 2 standard error of the mean in the last sign. digits.

Table 7: Selected results for intrinsic parameters calculated based on groundmass olivine-clinopyroxene-Ti-magnetite assemblages using QUIIF (Andersen *et al.*, 1993)

Assemblage	Olivine	Clinopyroxene	Spinel	Pressure (kbar)	Temperature (°C)		Oxygen fugacity		Silica activity	
					Absolute	Uncertainty	Absolute	ΔFMQ		Uncertainty
Groundmass	A19-4a-47 (micro; core)	A19-4a-2 (VI)	A19-4a-9 (5; core)	1	745	158	-11.2	4.7	3.4	-
Groundmass	A19-4a-48 (micro; mantle)	"	"	1	754	161	-11.1	4.6	3.4	-
Groundmass	A19-4c-146 (micro; core)	A19-4c-108 (VI)	A19-4c-88 (5; core)	1	832	156	-8.8	5.2	2.9	-
Groundmass	A19-4c-147 (micro; mantle)	A19-4c-122 (VI)	A19-4c-89 (5; core)	1	771	149	-10.2	5.1	3.1	-
Groundmass	A19-4c-151 (micro; rim)	"	"	1	809	117	-11.0	3.4	2.2	-
Groundmass	A19-06-25 (pheno; rim)	A19-06-34 (VI)	A19-06-48 (5; core)	1	743	182	-11.4	4.5	3.9	-
Groundmass	A19-06-23 (micro; rim)	A19-06-40 (VI)	A19-06-51 (5; rim)	1	775	166	-10.8	4.3	3.4	-

Pressure was fixed at 1 bar; silica activity could not be calculated (iteration limit).

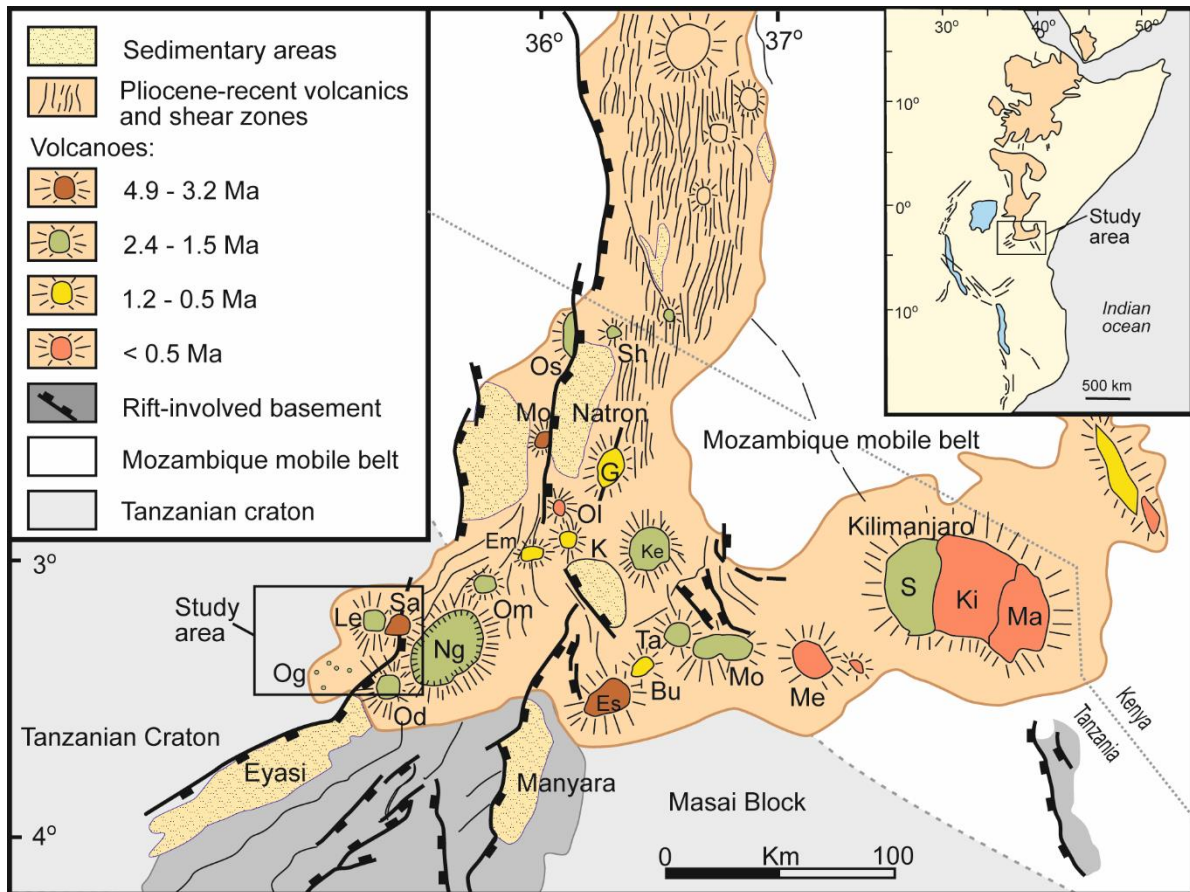


Fig. 1. Main structural and magmatic features in the North Tanzania Divergence zone (modified from Le Gall *et al.*, 2008). Volcanoes: Bu, Burko; Em, Embagai; Es, Essimigor; G, Gelai; K, Kerimasi; Ke., Ketumbeine; Ki, Kibo; Kw, Kwaraha; Le, Lemagarut; M, Monduli; Ma, Mawenzi; Me, Meru; Mo, Mosonik; Ng, Ngorongoro; Og, Ogoi; Os, Oldoinyo Shira, Ol, Oldoinyo Lengai; S, Shira; Sa, Sadiman, Sh, Shombole; Ta, Tarosero. Age data are from Mollel (2007), Dawson (2008, and references therein), Nonnotte *et al.* (2008), Mana *et al.* (2012; 2015) and Peirce & Mana (2018).

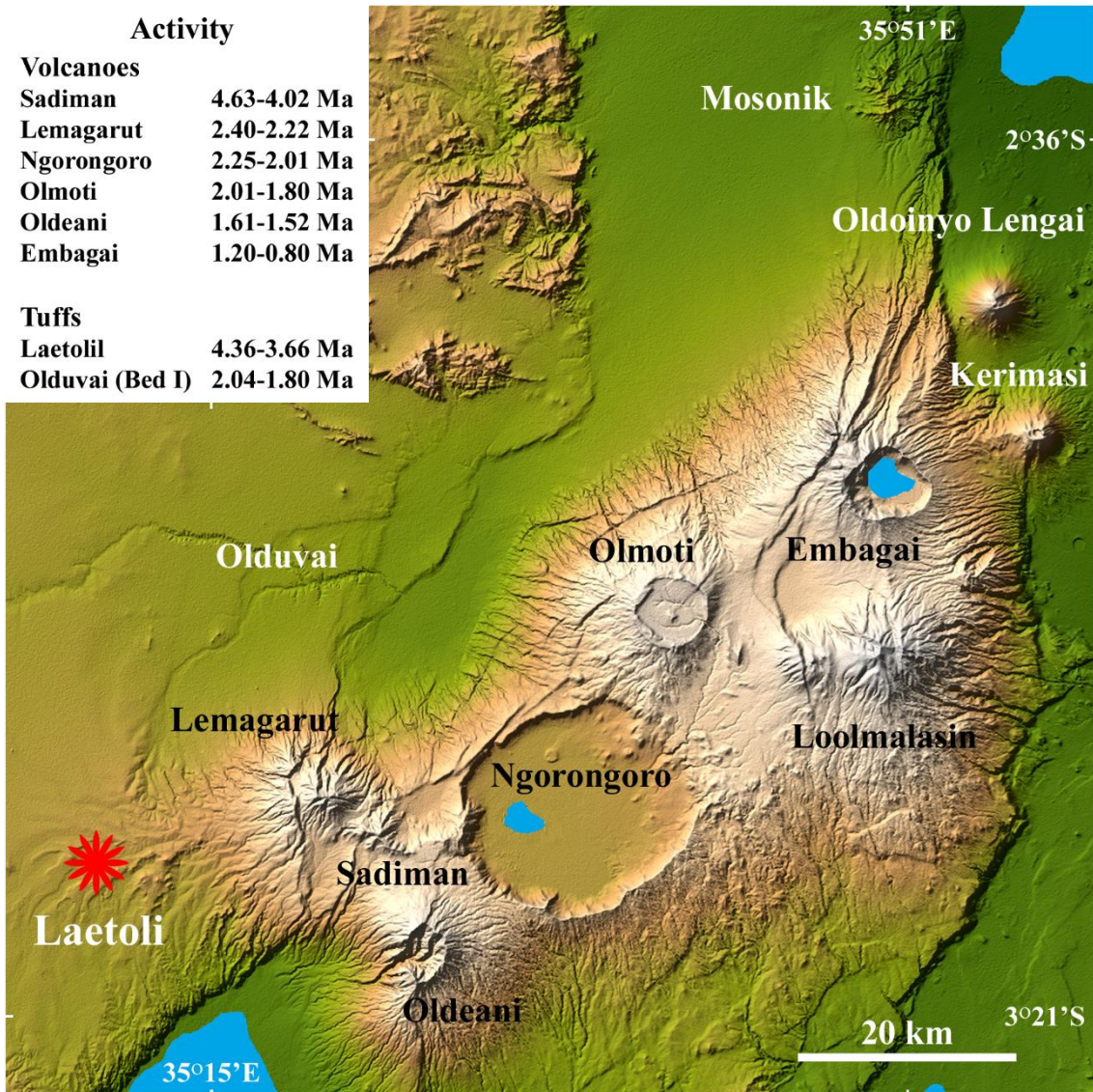


Fig. 2. Major volcanoes and tuffs in the Crater Highlands (shaded and coloured SRTM elevation model, February 2000). Courtesy NASA/JPL/NIMA. Age data are from Mollel (2007) and Deino (2011; 2012).

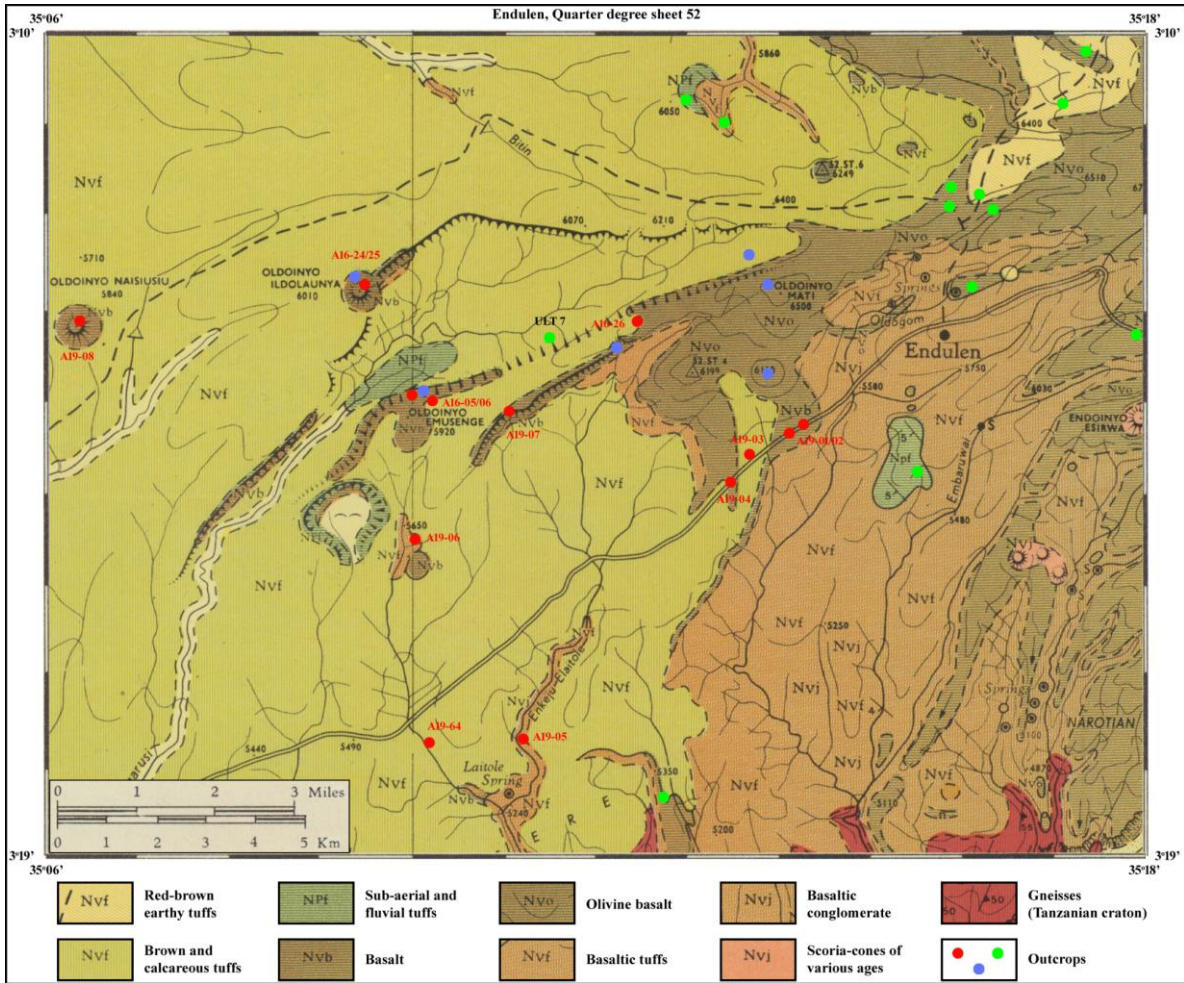


Fig. 3. Cropped Quarter degree sheet 52 Endulen (Pickering, 1964) showing geology of the Laetoli area. Outcrops with Ogot basalts are shown by red (this study) and blue (Mollel, 2007; Adelsberger *et al.*, 2011) dots; green dots are outcrops with Lemagarut basalts, Laetolil tuffs and tuffs with nephelinite xenoliths (likely to be from Sadiman).

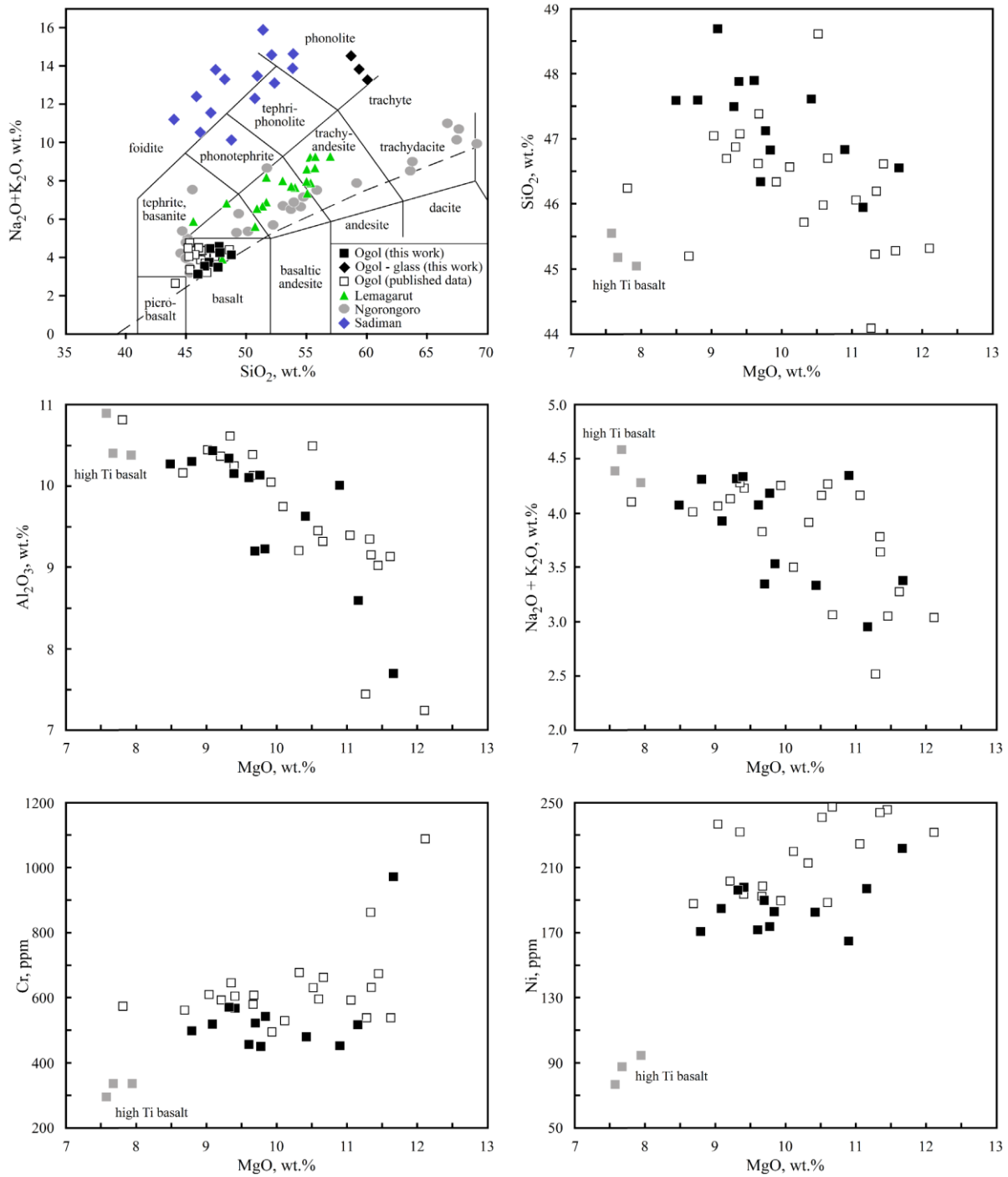


Fig. 4. Compositional variations for the Ogol basalts. (a) Total alkali vs SiO_2 diagram (TAS; Le Bas *et al.*, 1986). The dashed line divides alkaline and tholeiitic basalts. Compositions of basaltic series from the Lemagarut and Ngorongoro volcanoes and nephelinites from the Sadiman volcano are shown for comparison; (b-f) Major and compatible element variations. Published data are from Nonnotte (2007), Mollel (2007), Adelsberger *et al.* (2011), Mollel *et al.* (2011) and Zaitsev *et al.* (2012).

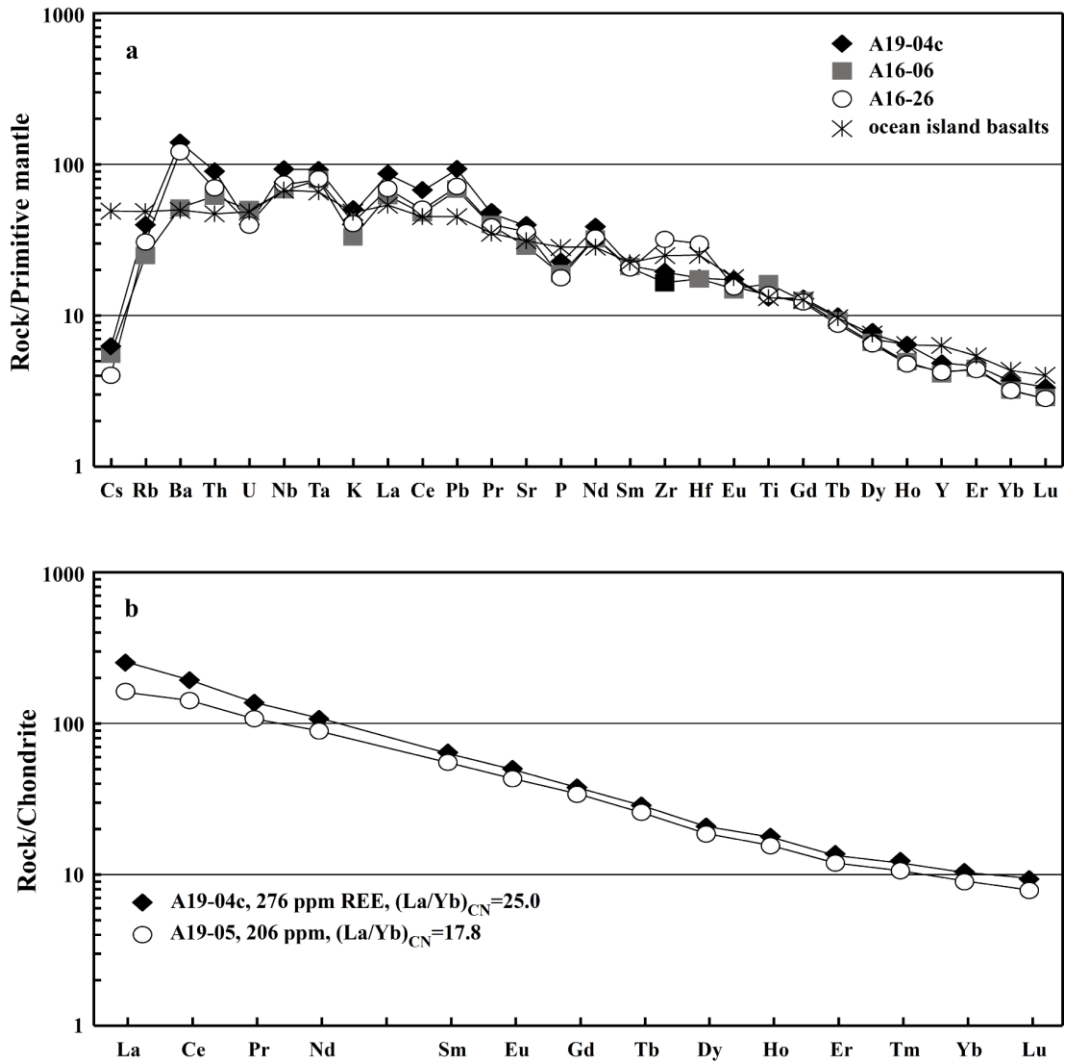


Fig. 5. Primitive mantle- and chondrite-normalised plots for selected Ogot basalt samples. Data for ocean island basalts, normalizing values for primary mantle and chondrite are from Sun & McDonough (1989). For Pb and Ce recommended normalizing values of 0.071 and 0.0079 ppm respectively were used (Sun & McDonough, 1989).

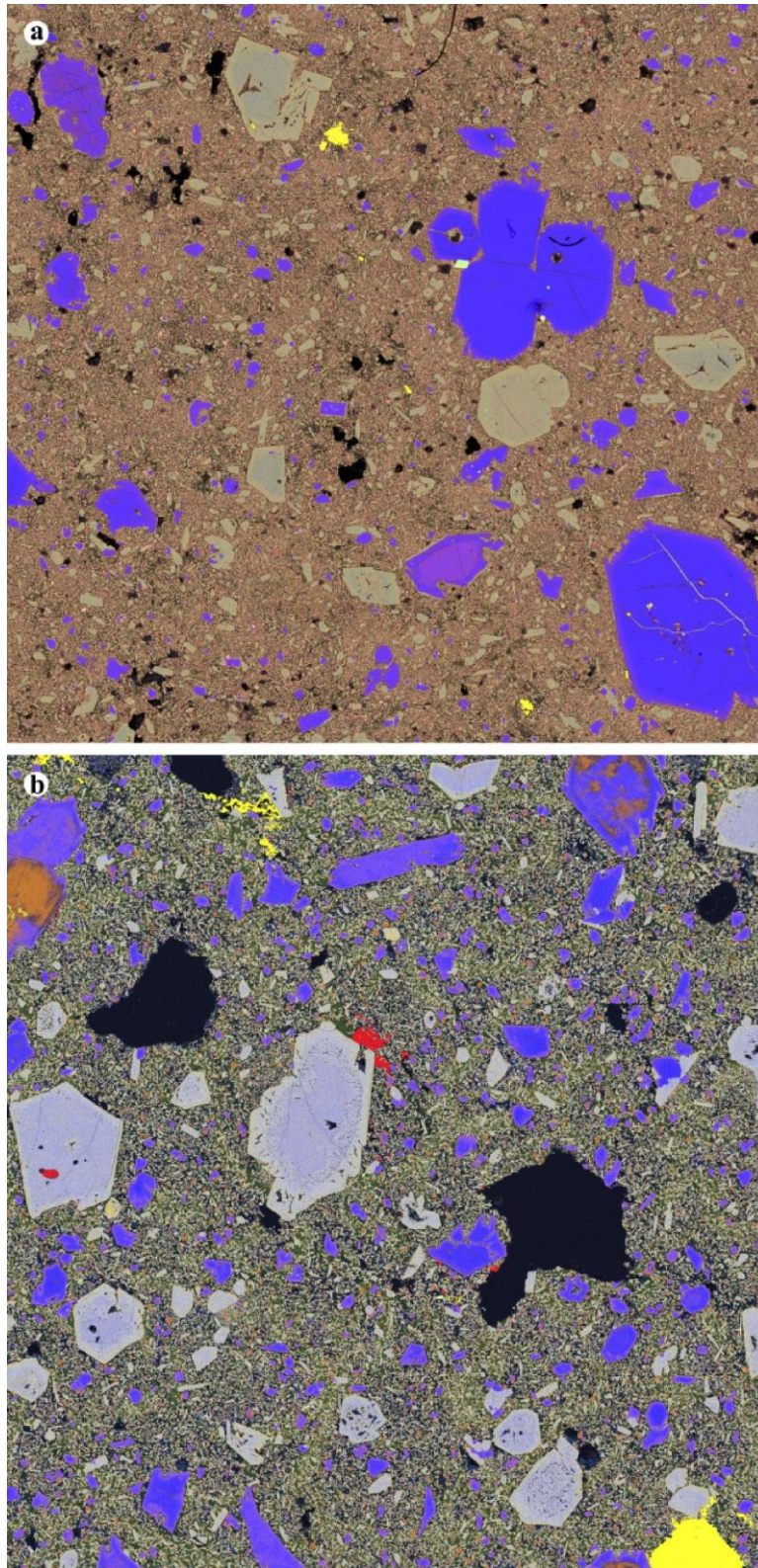


Fig. 6. False-coloured BSE montage maps of the Ogol basalts. (a) sample A16-05, field of view 9.48 x 9.22 mm, blue – olivine, beige – clinopyroxene, light green – magnesiochromite, pink in groundmass – magnetite, yellow – calcite, black – holes; (b) A16-26, field of view 9.48 x 10.20 mm, blue – olivine, orange-brown – olivine alteration, light violet to light gray – clinopyroxene, orange-brown in groundmass – magnetite, yellow – calcite, red – baryte.

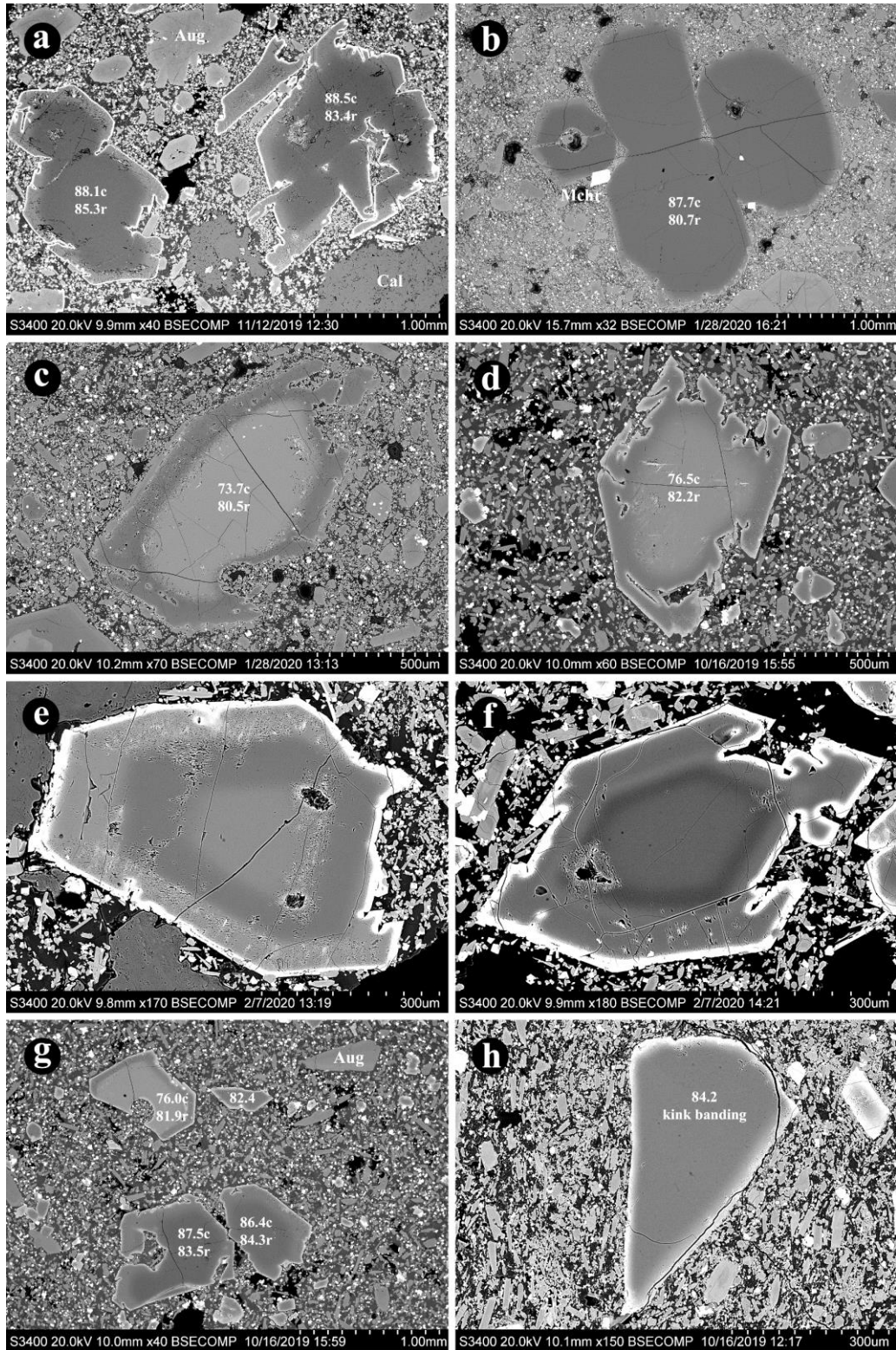


Fig. 7. Forsterite morphology and zonation. (a) hopper forsterite with normal zonation, A19-06, (b) euhedral to subhedral forsterite with normal zonation, Mchr – magnesiochromite, A16-05, (c) hopper forsterite with reverse zonation, A16-05, (d) hopper forsterite with reverse zonation, A19-07, (e) euhedral forsterite with repetitive zonation, A19-01, (f) hopper forsterite with repetitive zonation, A19-01, (g) forsterite with normal zonation, fragment of forsterite with reverse zonation and angular microphenocrystals (A19-07), (h) fragment of unzoned forsterite, A19-08. Numbers are Fo values (mol. %), c – core, r – rim. BSE images.

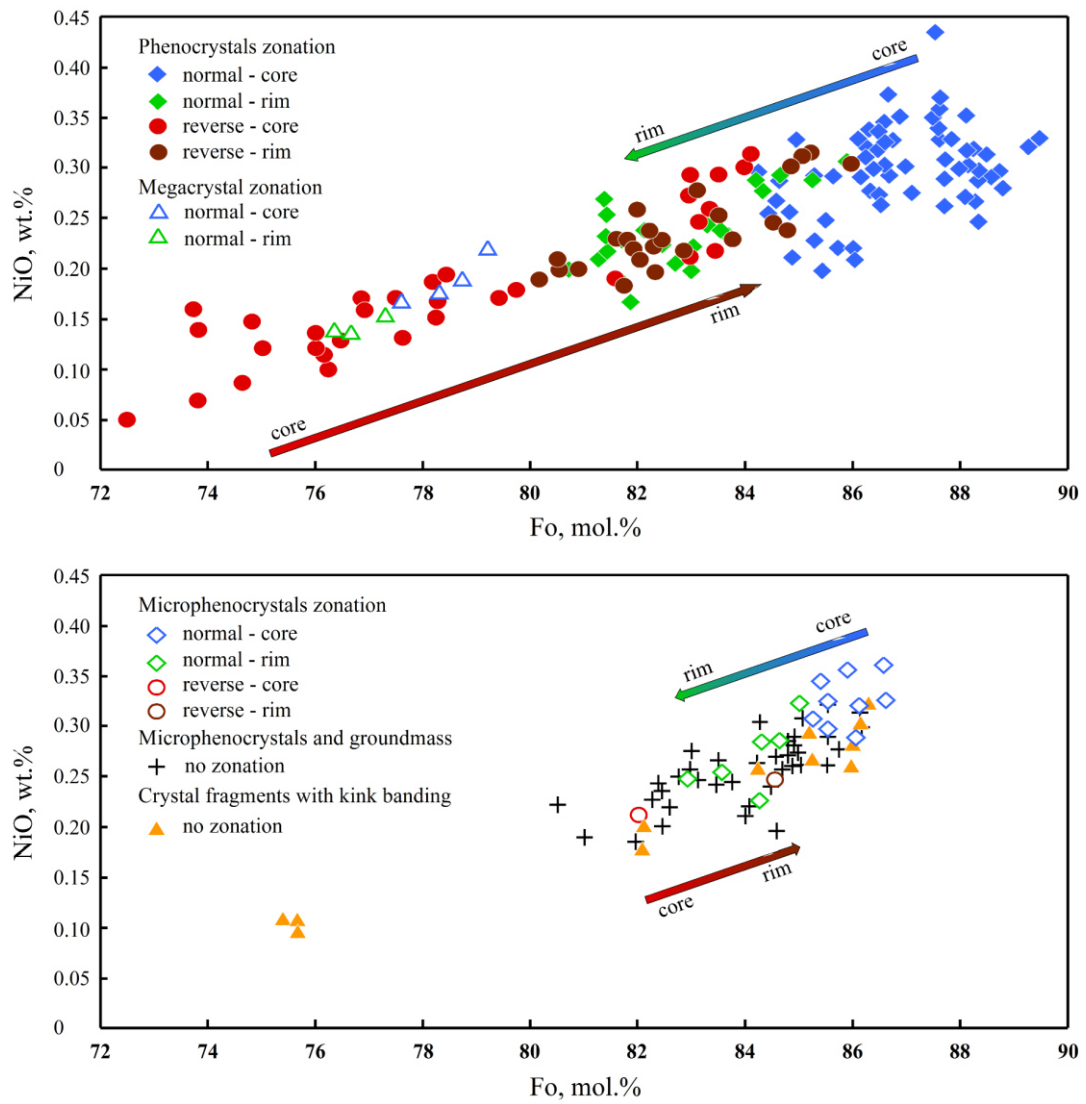


Fig. 8. Forsterite composition, Fo (mol. %) vs NiO (wt %) in the Ogol basalts.

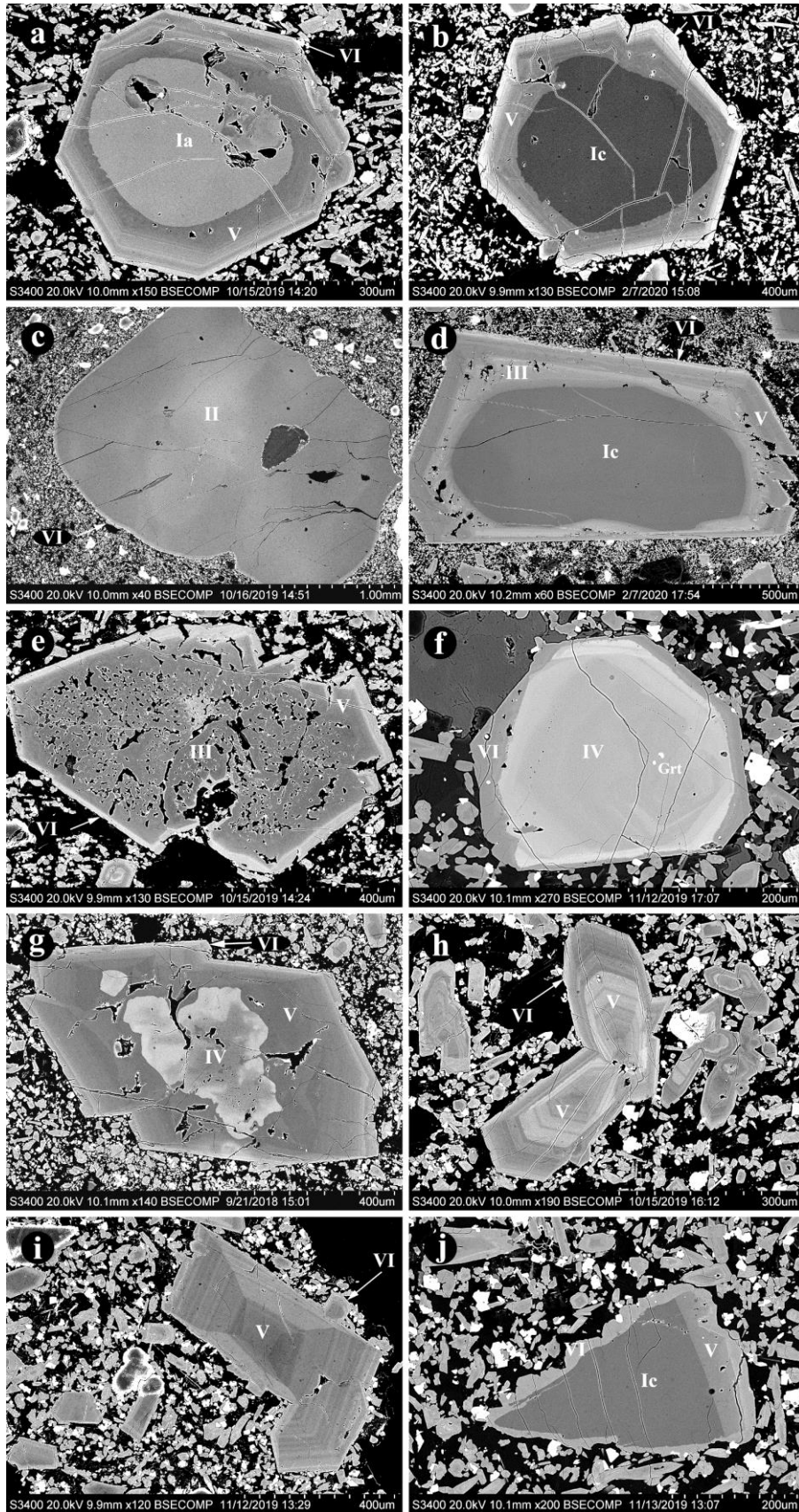


Fig. 9. Clinopyroxene morphology and zonation. I – VI – clinopyroxene population, (a, e) sample A19-64, (b) A19-01, (c) A19-08, (d) A19-02, (f) A19-04a, (g) A16-06, (h) A19-06, (i) A16-05, (j) A19-04b. Crt – schorlomite, BSE images.

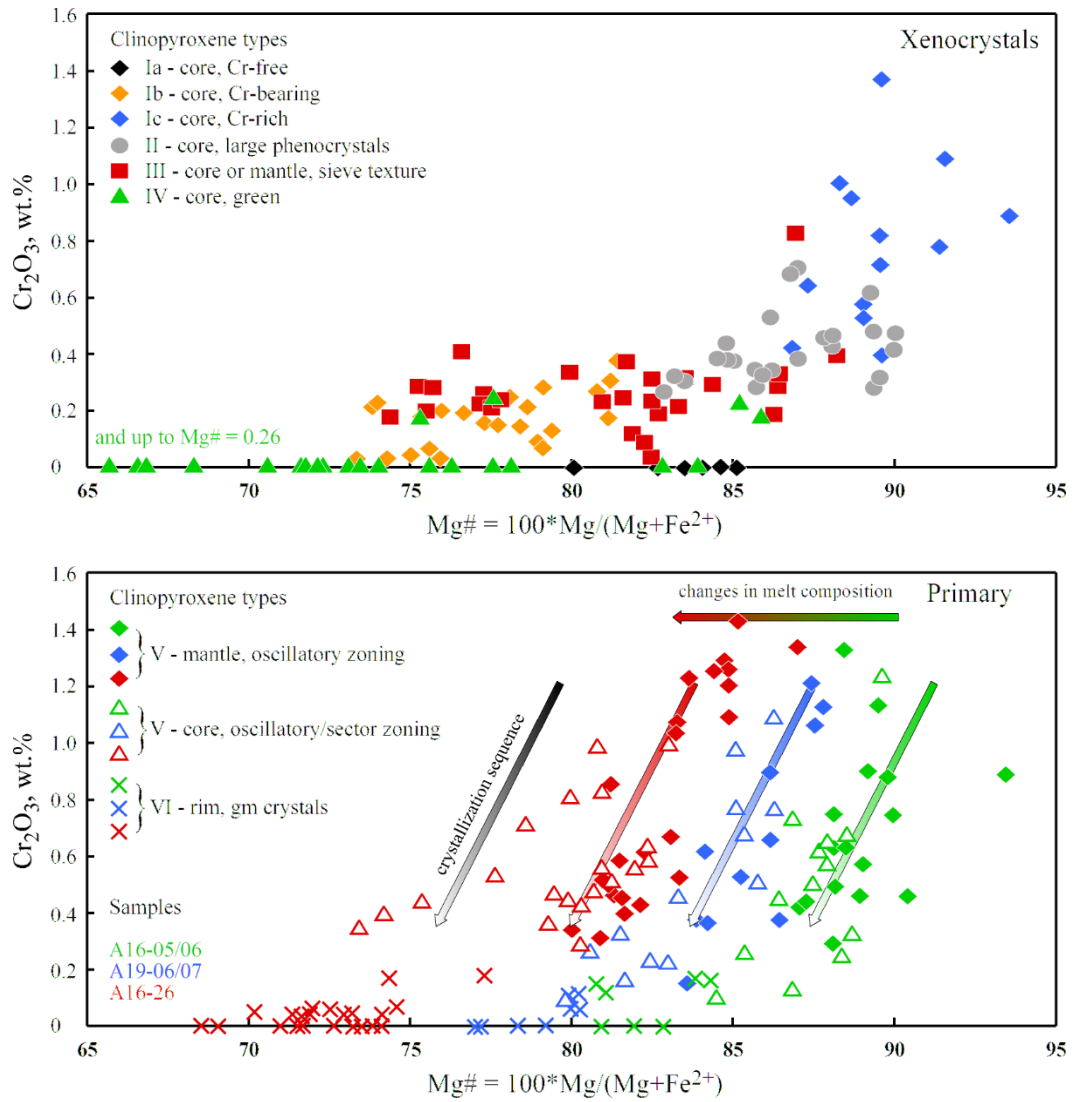


Fig. 10. Relationships between Mg\# ($100 \cdot \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$ from at %) and Cr_2O_3 (wt %) in clinopyroxene. Arrow “crystallization sequence” show compositional changes in population V (core to rim) and from population V to VI.

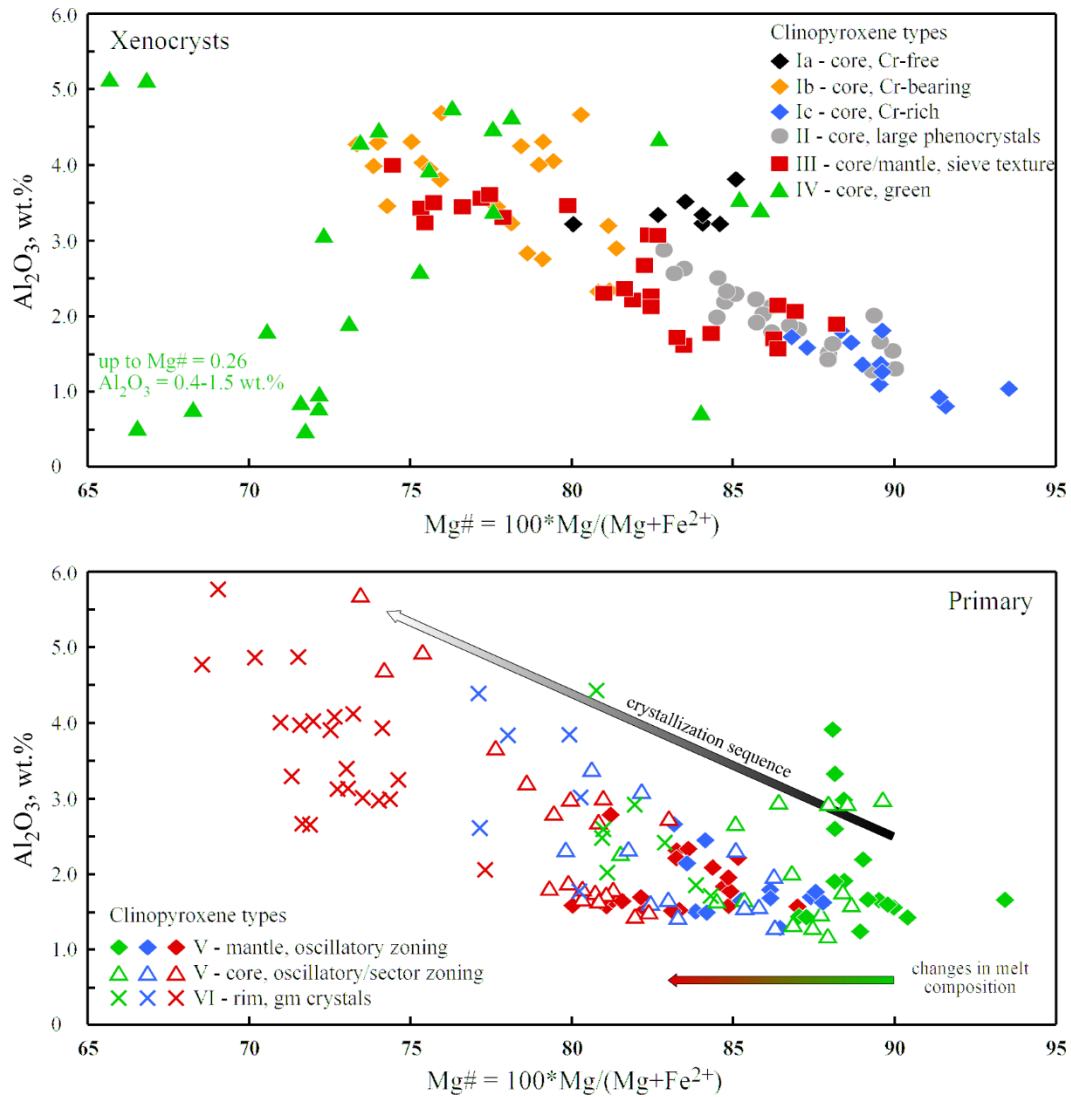


Fig. 11. Relationships between Mg# ($100 \cdot \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$ from at %) and Al_2O_3 (wt %) in clinopyroxene. Arrow “crystallization sequence” show compositional changes in population V (core to rim) and from population V to VI.

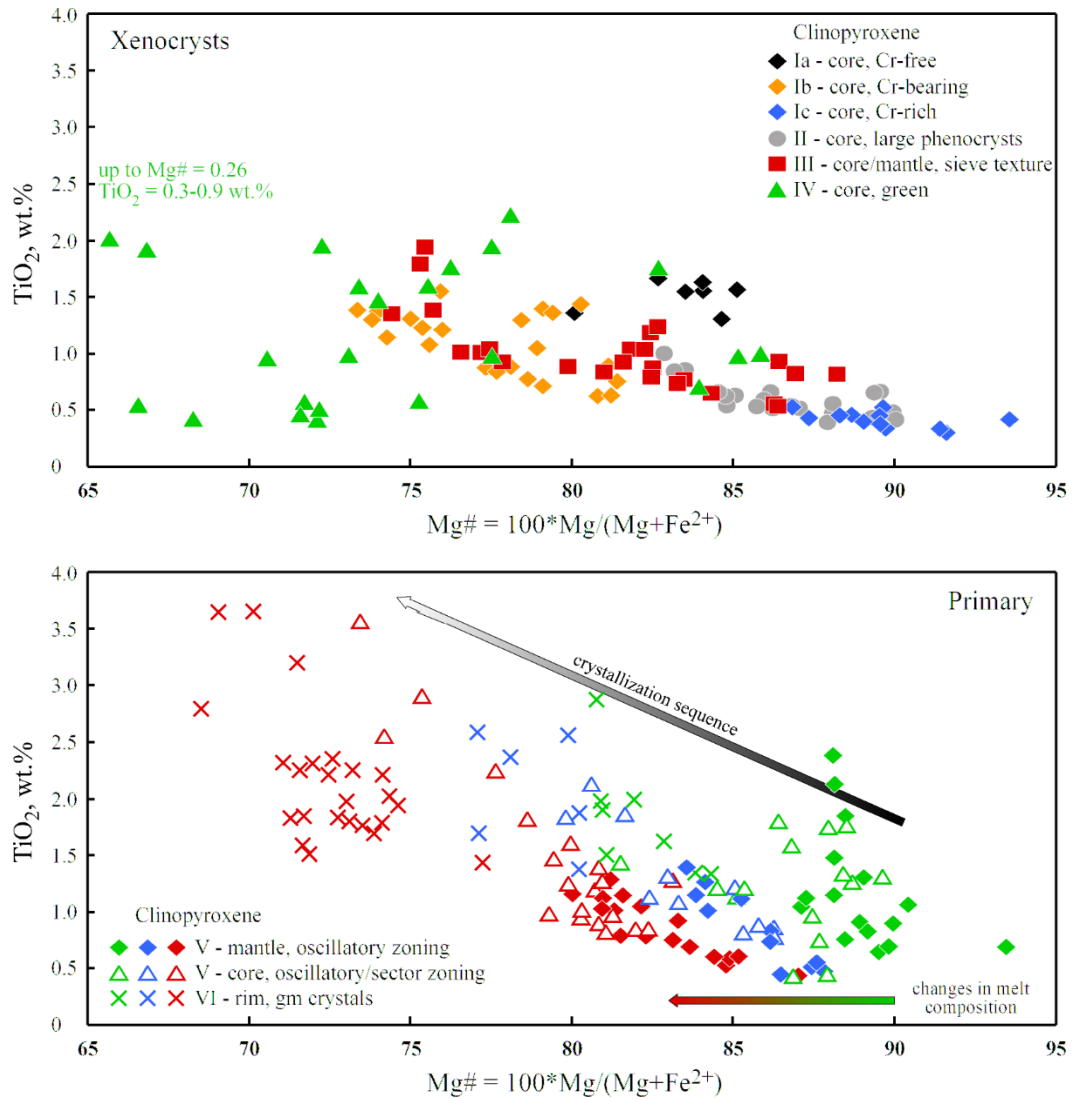


Fig. 12. Relationships between Mg# ($100 \cdot \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$ from at %) and TiO₂ (wt %) in clinopyroxene. Arrow “crystallization sequence” show compositional changes in population V (core to rim) and from population V to VI.

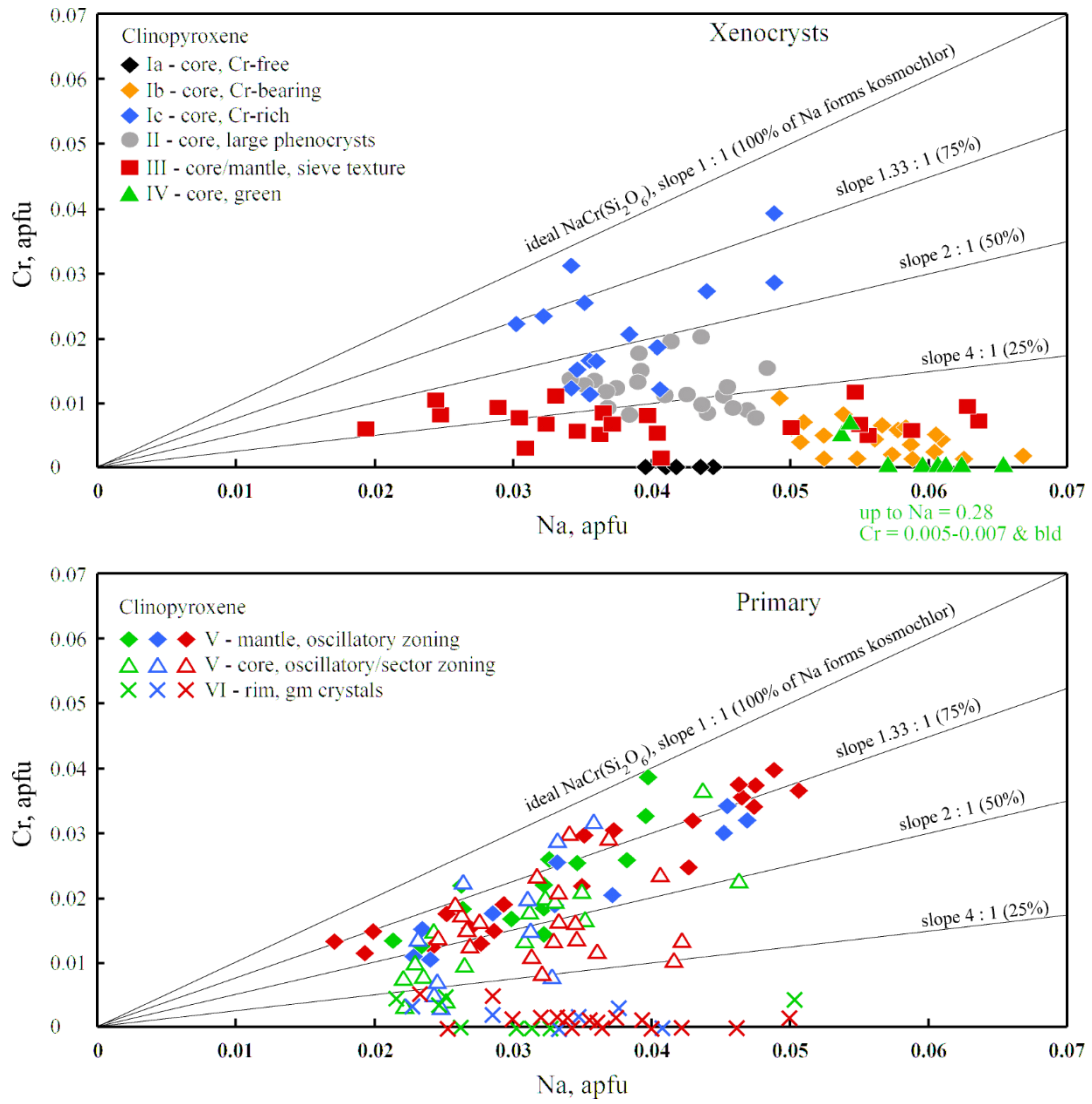


Fig. 13. Relationships between Na and Cr (apfu) in clinopyroxene.

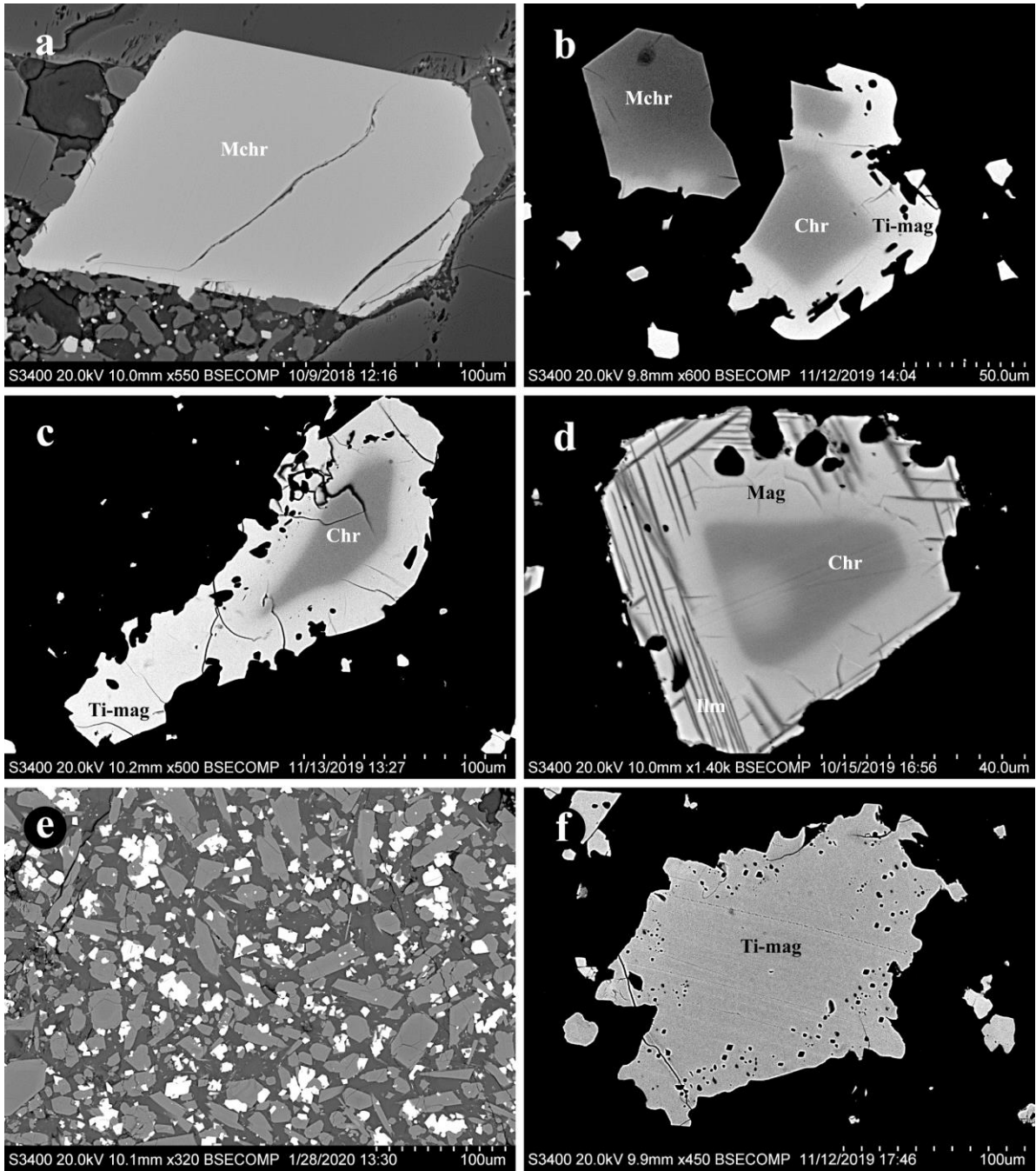


Fig. 14. Morphology and zonation of spinel-group minerals. (a) euhedral magnesiochromite (Mchr) at contact forsterite phenocrystal and groundmass, A16-05; (b) euhedral magnesiochromite in forsterite phenocrystals and zoned chromite (Chr) - magnetite-ulvöspinel (Ti-mag) at contact forsterite phenocrystal and groundmass, A19-06, (c) zoned chromite (Chr) - magnetite-ulvöspinel (Ti-mag) microphenocrystal, A19-04b, (d) zoned chromite (Chr) - magnetite (Mag) microphenocrystals with ilmenite exsolution (Ilm) in magnetite, A19-64, (e) euhedral to subhedral magnetite-ulvöspinel (white) in groundmass, A16-05, (f) unhedral magnetite-ulvöspinel xenocrystal, A19-04a. BSE images.

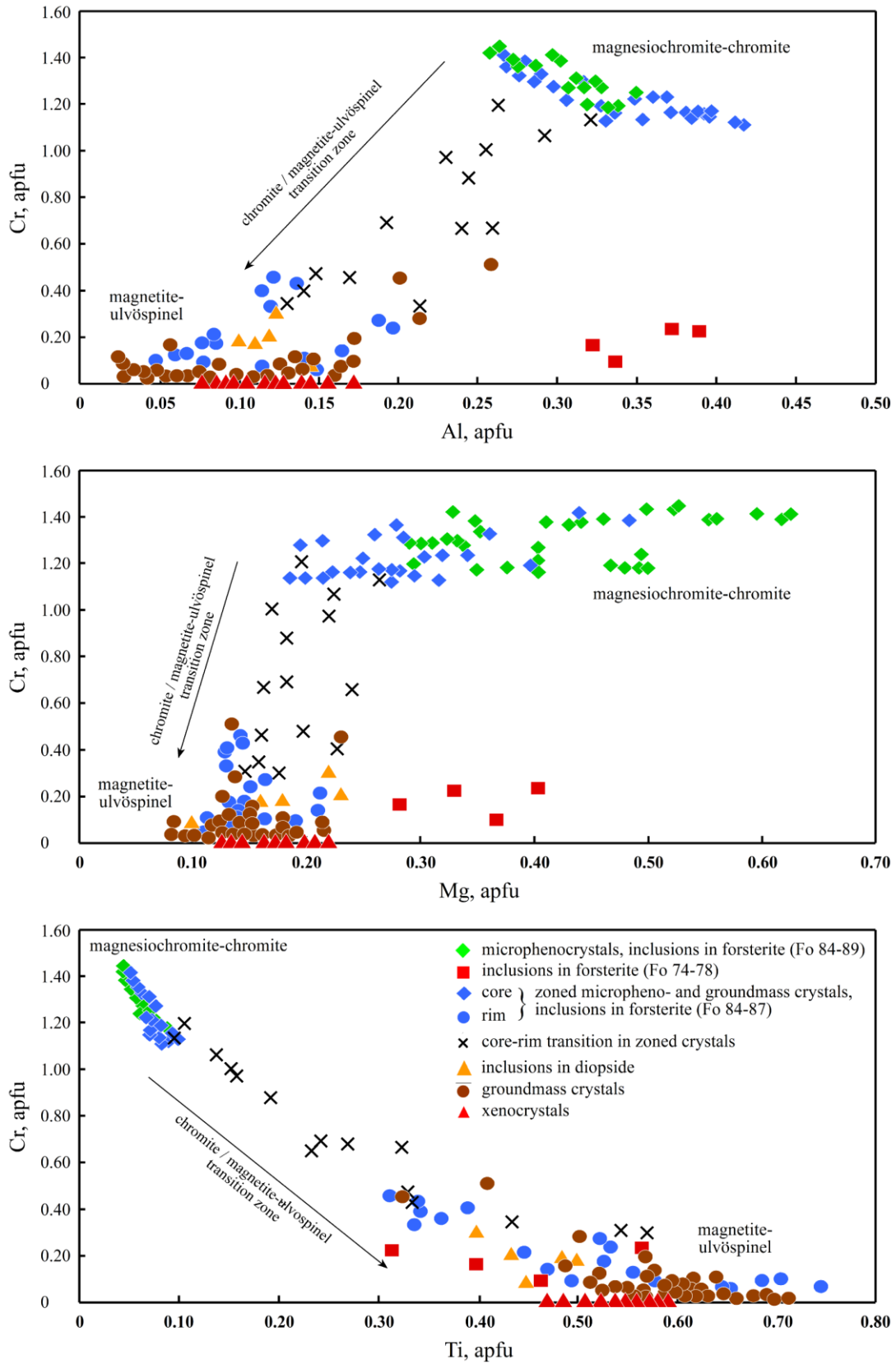


Fig. 15. Compositional variations of spinel-group minerals.

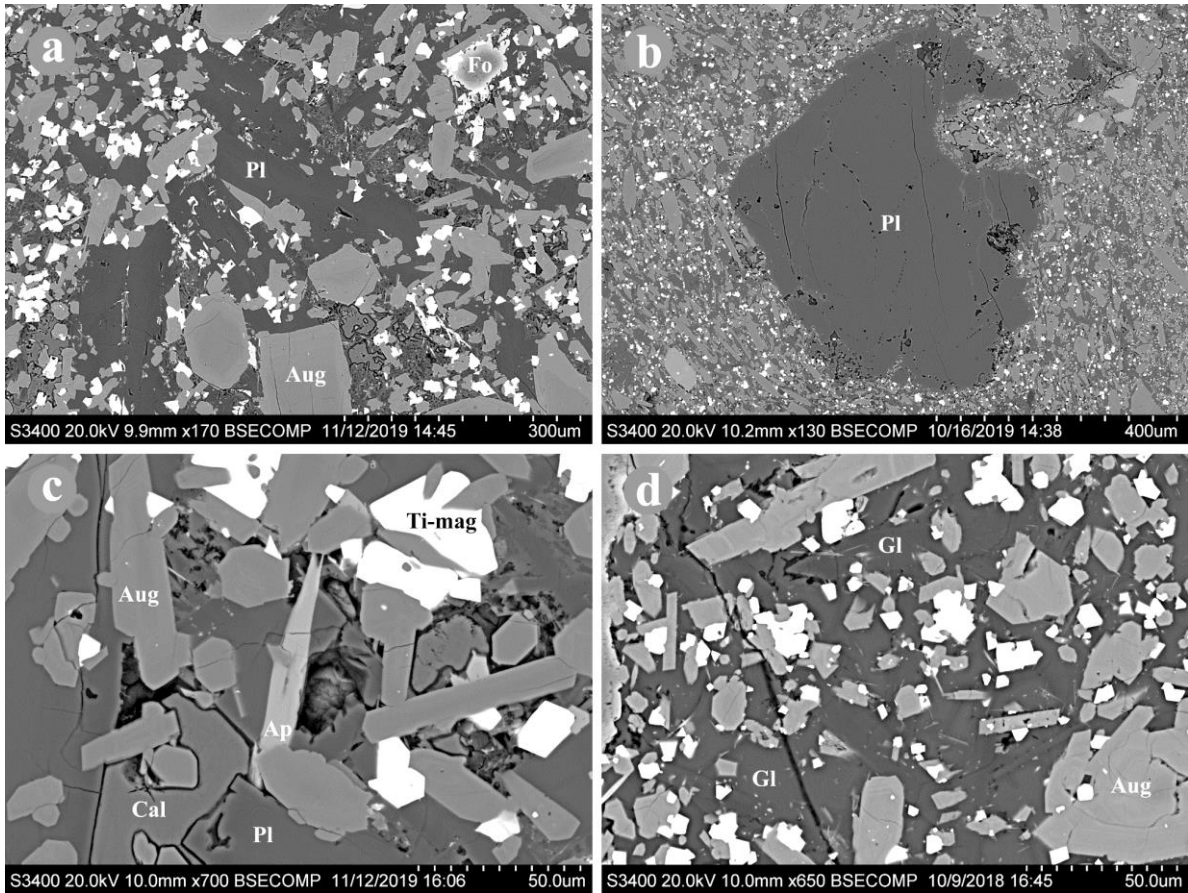


Fig. 16. Microphenocrysts and groundmass minerals. (a) andesine (Pl) microphenocrystals, A19-06, (b) oligoclase (Pl) xenocrystal, A19-08, (c) apatite (Ap) microphenocrystal, A19-07, (d) glass (Gl) with apatite microlites (gray), A16-06. Fo forsterite, Aug augite, Ti-mag magnetite-ulvöspinel, Cal calcite. BSE images.

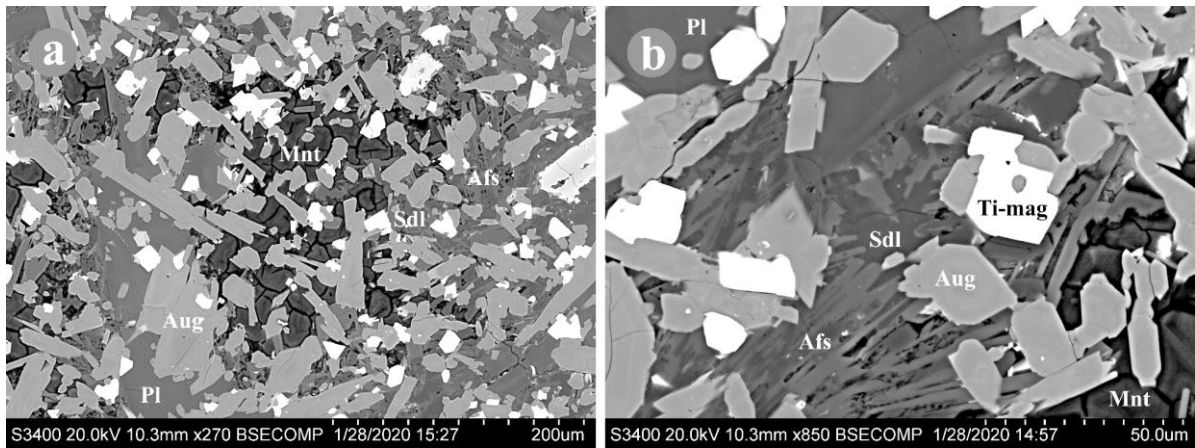


Fig. 17. Glass alteration to montmorillonite (Mnt), K-Na feldspar (Afs), Na zeolite (not shown) and sodalite (Sdl). Pl andesine, Aug augite, Ti mag magnetite-ulvöspinel_{ss}, A16-26. BSE images.

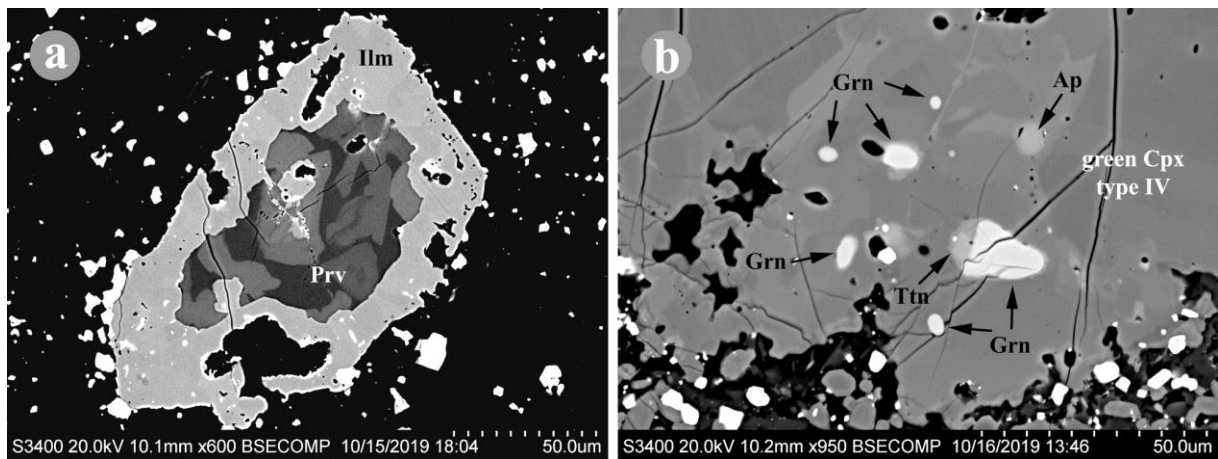


Fig. 18. Xenocrysts in the Ogot basalts. (a) perovskite (Prv) with ilmenite (Ilm) reaction rim, white inclusions in ilmenite are REE-rich minerals, (b) schorlomite (Grn), titanite (Ttn) and apatite (Ap) in green clinopyroxene of type IV, A19-08. BSE images.

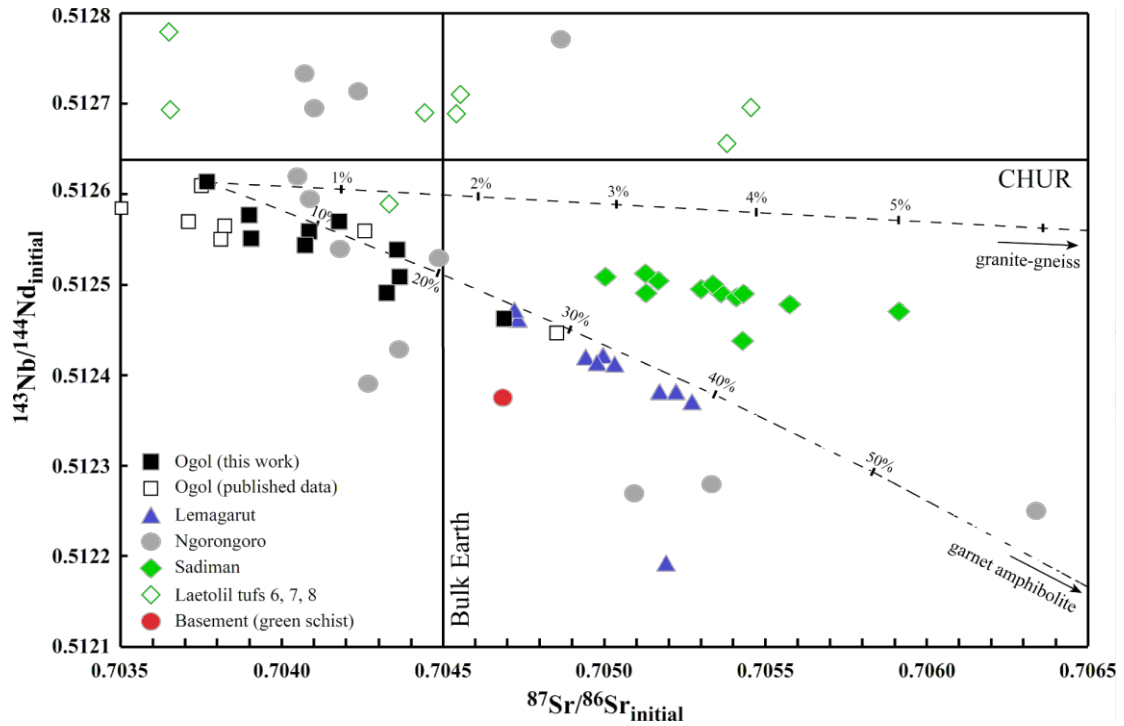


Fig. 19. Sr and Nd initial ratios (2.3 Ma) for the Ogol basalts, Lemagarut, Ngorongoro and Sadiman volcanoes, Laetolil tuffs and green schist from the Tanzania Craton. Data are from this study and Paslick *et al.* (1995, 1996), Mollel (2007), Nonnotte (2007), Adelsberger *et al.* (2011) and Zaitsev *et al.* (2019). Dashed lines show mixing isotope models between Ogol basalt (A16-05), granite-gneiss (A16-12) and garnet amphibolite (A16-16c).

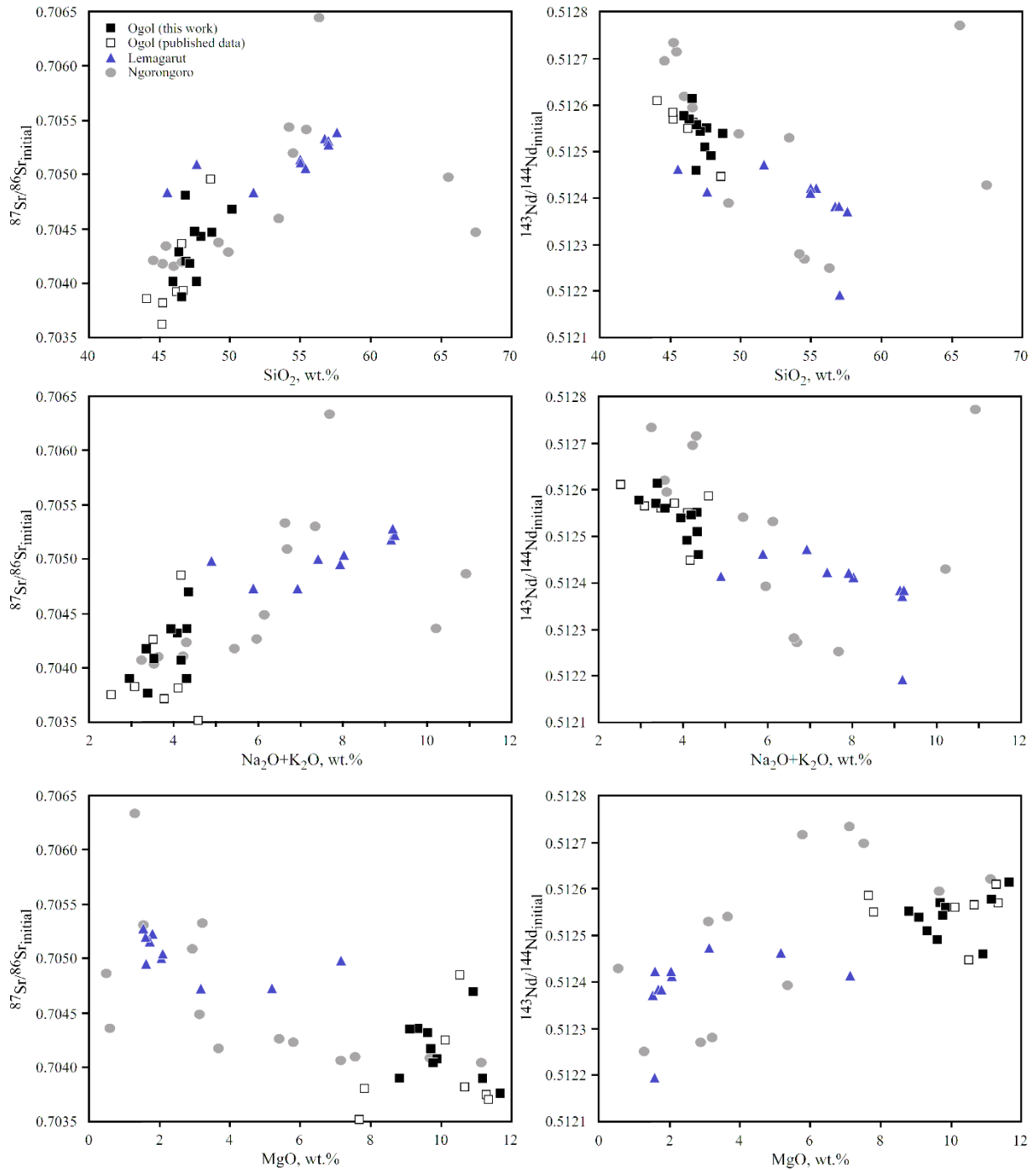


Fig. 20. Relationships between Sr / Nd initial ratios and major elements for the Ogot basalts and Lemagarut and Ngorongoro volcanoes. Data are from this study and Mollet (2007), Nonnotte (2007) and Adelsberger *et al.* (2011).

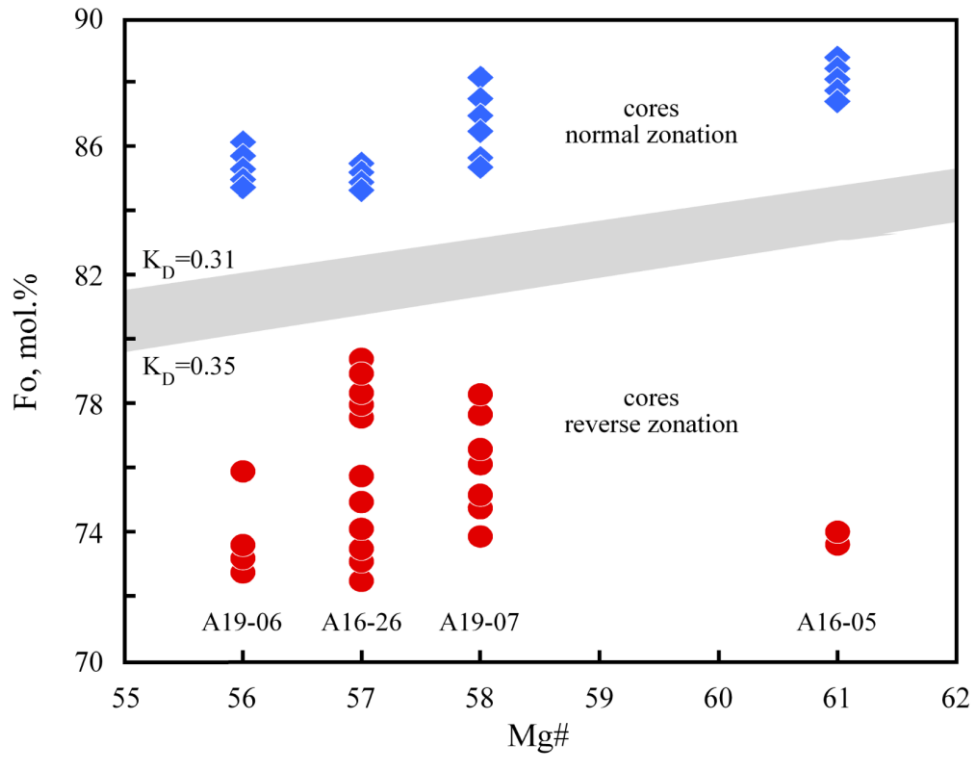


Fig. 21. Whole rocks Mg# ($100 \cdot \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$) vs forsterite composition. Only cores of phenocrysts with normal and reverse zonation were used for the plot. The gray field represent olivine/melt equilibrium compositions for $K_D = 0.31$ - 0.35 (Ulmer, 1989).

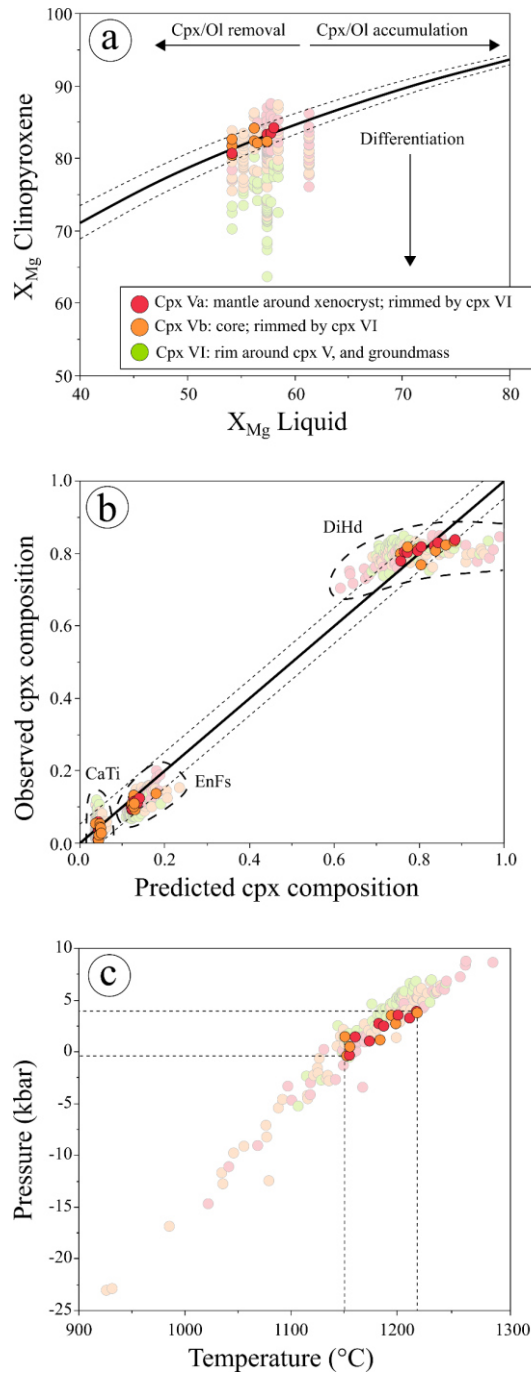


Fig. 22. Clinopyroxene-liquid thermobarometry for the Ogol basalts according to Neave & Putirka (2017), paired with the temperature equation 33 from Putirka (2008). The whole rock composition was used as a proxy for the coexisting liquid to clinopyroxene. (a) Rhodes diagram (X_{Mg} clinopyroxene vs X_{Mg} liquid) for the studied basalts. X_{Mg} is the molar $Mg/[Mg+Fe_{tot}] \cdot 100$. The curves illustrate the equilibrium interval between clinopyroxene and liquid by assuming an exchange coefficient $K_{D(Fe-Mg)}^{cpx-liq}$ of 0.27 ± 0.3 ; (b) Observed clinopyroxene compositions vs predicted clinopyroxene compositions including diopside-hedenbergite (DiHd), enstatite-ferrosilite (EnFs) and Ti-Tschermak's clinopyroxene (CaTi; Putirka, 2008). The accepted range of equilibrium is indicated by the dashed lines ($\pm 5\%$, absolute). The predicted DiHd component sometimes exceeds 1.0 (up to 2.3), with the corresponding pressure calculations giving extraordinary low values < -5 kbar; (c) Pressure-temperature correlation of the applied thermobarometer. Highlighted are the results that fulfill the equilibrium constraints.

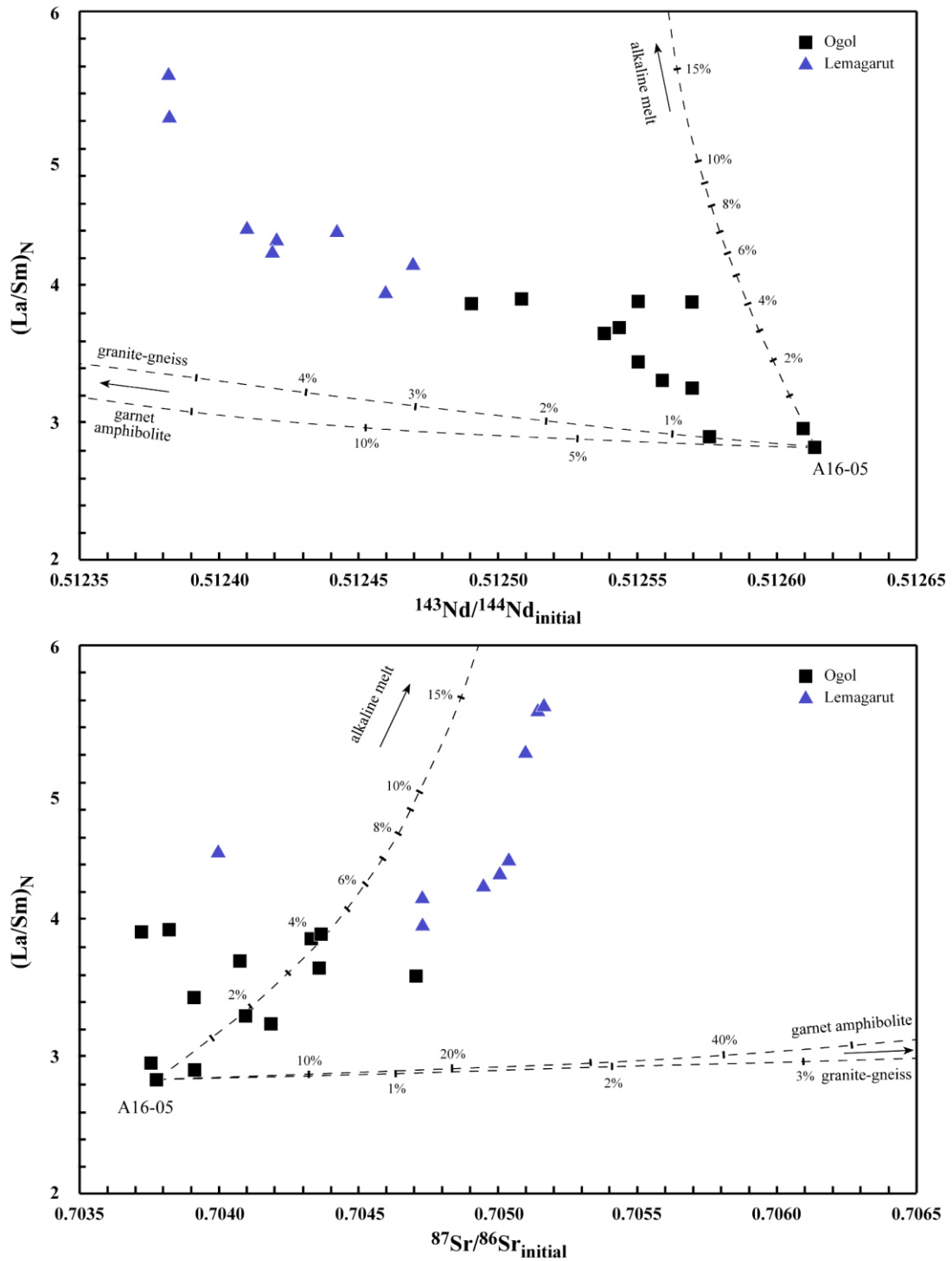


Fig. 23. Nd and Sr initial isotope ratios vs $(La/Sm)_N$. Dashed lines show variations of the Ogol primitive basalt (A16-05) during fractional crystallization with crustal contamination of Precambrian granite-gneiss (A16-12), garnet amphibolite (A16-15c) or mixing with alkaline melt (sample Sad-10 from Zaitsev *et al.*, 2012). AFC model, $R = 0.8$ (ratio of mass assimilation rate to the fractional crystallization rate).

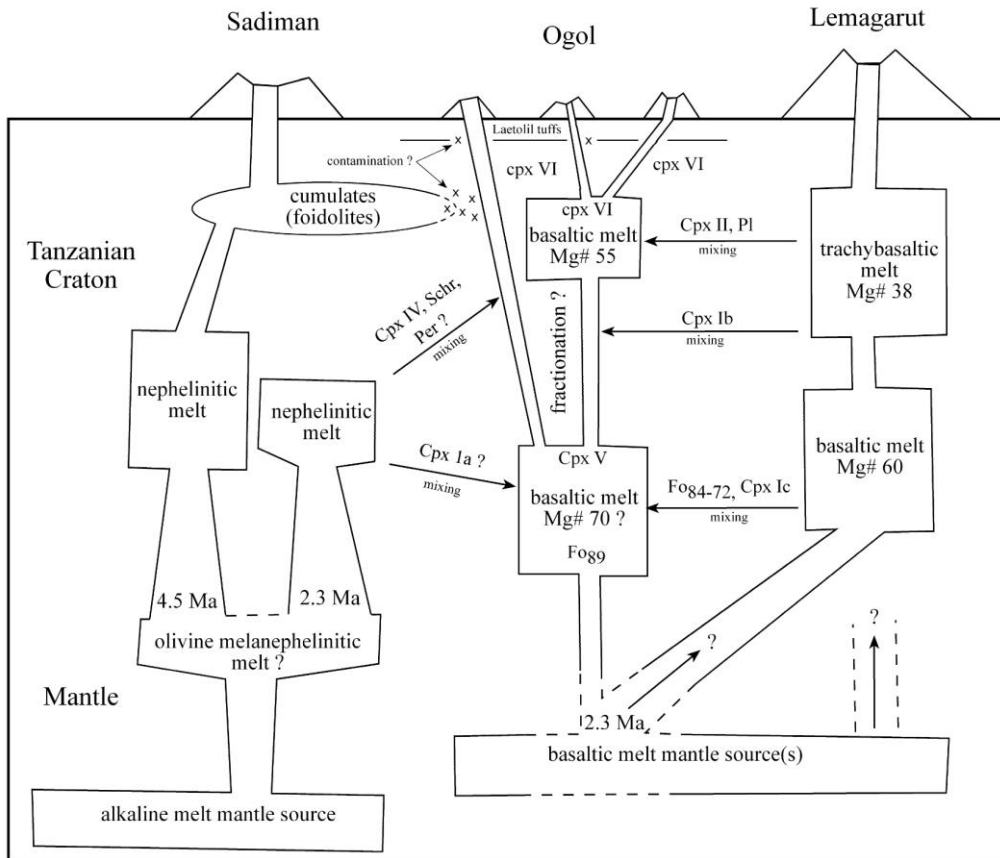


Fig. 24. Schematic diagram showing possible relationships between magmatic systems of Ogol, Lemagarut and Sadiman. The horizontal and vertical sketches are not to scale.

Appendix II

Submitted manuscript

Study 2

Braunger, S., Marks, M. A. W., Zaitsev, A. N., Wenzel, T., and Markl, G., submitted.
The Petrology of the Tarosero Volcanic Complex: Constraints on the formation of
extrusive agpaitic rocks. *Journal of Petrology*.

1 **The Petrology of the Tarosero Volcanic Complex:**

2 **Constraints on the formation of extrusive agpaitic rocks**

3

4 Braunger, S.^{1*}, Marks, M.A.W.¹, Wenzel, T.¹, Zaitsev, A.N.^{2,3} and Markl, G.¹

5 ¹Department of Geosciences, Eberhard Karls Universität Tübingen, Wilhelmstr. 56, D-72074 Tübingen,
6 Germany

7 ²Department of Mineralogy, St. Petersburg State University, University Emb. 7/9, St. Petersburg,
8 199034, Russia

9 ³Imaging and Analysis Centre, Department of Earth Sciences, Natural History Museum, Cromwell
10 Road, London, SW7 5BD, UK

11 * = corresponding author: Email: simonbraunger@hotmail.com; Telephone: +49-(0)7071-29-73155;
12 Fax: +49-(0)7071-29-3060.

13

14

15 **ABSTRACT**

16 The Quaternary Tarosero volcano is situated in the East African Rift of northern Tanzania and mainly
17 consists of trachyte lavas and some trachytic tuffs. In addition, there are minor occurrences of extrusive
18 basalts, andesites, latites, as well as peralkaline trachytes, olivine trachytes and phonolites. Some of the
19 peralkaline phonolites contain interstitial eudialyte, making Tarosero one of the few known occurrences
20 for extrusive agpaitic rocks. This study investigates the genetic relationships between the various rock
21 types and focuses on the peculiar formation conditions of the extrusive agpaitic rocks using a
22 combination of whole-rock geochemistry, mineral chemistry, petrography, thermodynamic calculations
23 and fractional crystallization models. The Tarosero rocks derive from an amphibole- and phlogopite-
24 bearing mantle source and formed at redox conditions around or below the fayalite-magnetite-quartz
25 buffer (FMQ). During multi-level magmatic fractionation at depths between ~35 km and the shallow

26 crust, temperature decreased from > 1100 °C at near-liquidus conditions in the basalts to ~ 700 °C in
27 the peralkaline residue. The andesite shows a considerably lower silica activity of $a_{\text{SiO}_2}=0.15$ and a low
28 molar Na/K ratio of 1.5 compared to the other intermediate rock types ($a_{\text{SiO}_2}\geq 0.4$; Na/K=2.5-3.0),
29 indicating a different parental magma source. The genetic relationships between the primitive basalts
30 and the intermediate latites cannot be explained by simple fractional crystallization. However, latites
31 evolved towards the trachytes by high-pressure fractionation of clinopyroxene, magnetite and olivine.
32 Further low-pressure fractional crystallization of plagioclase and alkali feldspar, with minor
33 contributions of clinopyroxene, amphibole, magnetite and ilmenite, resulted in the formation of the
34 peralkaline trachytes and phonolites. Fractionation towards the peralkaline rocks was probably
35 accompanied by minor crustal contamination. In general, agpaitic magmas evolve under closed system
36 conditions which impedes the escape of volatile phases. For the extrusive rocks at Tarosero, exsolution
37 of fluids and halogens was prevented by a low water activity due to the early crystallization of OH-
38 bearing amphibole. This resulted in high concentrations of Rare Earth Elements (REE), High Field
39 Strength Elements (HFSE) and the formation of eudialyte in the most evolved peralkaline phonolites.
40 In contrast, the peralkaline olivine trachytes contain olivine instead of amphibole phenocrysts,
41 demonstrating a derivation from the latites by fractional crystallization (plagioclase, clinopyroxene,
42 magnetite, olivine) and assimilation at comparably low-pressure conditions. Due to the lack of a mineral
43 phase acting as a water sink, exsolution of fluids and loss of water-soluble elements prevented an
44 enrichment of REE, HFSE and the final crystallization of eudialyte.

45

46 **KEY WORDS**

47 Eudialyte; extrusive agpaitic rocks; fractional crystallization; peralkaline; Tarosero

48

49

50 INTRODUCTION

51 Peralkaline rocks mainly occur in rift-related intraplate settings, and are defined by a molar (Na+K)/Al
52 ratio > 1 (e.g. Le Maitre *et al.*, 2002, Frost & Frost, 2008). Although they represent a volumetrically
53 minor group of igneous rocks with both, silica undersaturated and silica oversaturated varieties (e.g.
54 Dawson, 1998, Markl *et al.*, 2001, Marks *et al.*, 2003, Macdonald & Scaillet, 2006, Mann *et al.*, 2006,
55 Ren *et al.*, 2006, Ridolfi *et al.*, 2006, Zaitsev *et al.*, 2012, Estrade *et al.*, 2014, Cucciniello *et al.*, 2017),
56 peralkaline rocks are of economic and scientific interest due to their high abundances of incompatible
57 elements, including High Field Strength Elements (HFSE), Rare Earth Elements (REE) and volatiles
58 (e.g. Kogarko, 1990, Sørensen, 1992, Bailey *et al.*, 2001, Kalashnikov *et al.*, 2016, Smith *et al.*, 2016).
59 Most peralkaline rocks form by partial melting of a metasomatically enriched mantle source (e.g. Fitton
60 & Upton, 1987, Martin *et al.*, 1994, Upton *et al.*, 2003, Kogarko *et al.*, 2010, Markl *et al.*, 2010),
61 followed by subsequent intense fractional crystallization under variable redox conditions (see details in
62 Marks & Markl, 2017 and references therein).

63 Among the peralkaline rocks, the more common miaskitic types contain well-known HFSE-bearing
64 minerals, such as zircon, baddeleyite, titanite or perovskite, the relatively rare agpaitic rocks, however,
65 contain Na-Ca-HFSE-bearing and halogen-rich minerals such as eudialyte or wöhlerite instead (e.g.
66 Sørensen, 1997, Marks *et al.*, 2011, Marks & Markl, 2017). Agpaitic rocks belong to the most evolved
67 and most reduced alkaline rocks at all (Marks & Markl, 2017). A major prerequisite for the formation
68 of agpaitic rocks is the retention of volatiles in the melt (e.g. Marks & Markl, 2001, Marks & Markl,
69 2015). This is most likely reached in closed systems (Sørensen, 1992) and is in accordance with the fact
70 that most of the about 100 occurrences of agpaitic rocks are plutonic. However, the relatively few
71 extrusive agpaitic rocks (e.g. at Oktyabr'sky, Tarosero and Kontozero; Krivdik & Tkachuk, 1988,
72 Dawson, 1997, Petrovsky *et al.*, 2012) raise questions on how halogens and other volatiles can be
73 prevented from escaping to the atmosphere within a volcanic system.

74 The Tarosero Volcanic Complex in northern Tanzania is a well-suited locality to investigate such
75 formation processes as it comprises variably evolved alkaline to peralkaline extrusive rocks, including
76 rare and highly evolved agpaitic phonolites (Dawson, 2008). To date, no detailed study exists about the

77 mineralogy and petrology of this volcano. We present a detailed mineralogical, mineral chemical and
78 geochemical investigation of the Tarosero rocks. Combined with thermodynamic calculations of the
79 prevailing crystallization conditions and fractional crystallization models, we decipher the genetic
80 relationships between the various rock types, and suggest processes to form the highly evolved, agpaitic
81 extrusive rocks in the volcanic environment at Tarosero.

82

83 **GEOLOGY**

84 The Tarosero volcano in northern Tanzania is situated in the eastern branch of the East African Rift
85 valley, with the Kenyan and Tanzanian parts also known as the Gregory Rift. Rifting was initiated by
86 plume activity beneath Ethiopia 30 Ma ago and proceeded southwards along Pan-African suture zones
87 (e.g. Hofmann *et al.*, 1997 and references therein, Chorowicz, 2005). Volcanic magmatism in northern
88 Tanzania is linked to the North Tanzanian Divergence Zone (Fig. 1a; Le Gall *et al.*, 2008), which formed
89 along pre-existing zones of crustal weakness between the Tanzanian craton to the west and the
90 Mozambique mobile belt to the east (Smith, 1994).

91 Magmatism can be divided into two groups based on major faulting events (Dawson, 1992, Dawson,
92 2008): a pre-1.2 Ma group of mainly alkali basalt-phonolite (e.g. Tarosero, Monduli, Ngorongoro,
93 Oldoinyo Sambu, Ketumbeine, Kilimanjaro) and rather rare nephelinite±carbonatite volcanoes (e.g.
94 Burko, Essimigor, Mosonik), and a post-1.2 Ma group comprising volcanoes essentially of the
95 nephelinite-phonolite±carbonatite association (e.g. Meru, Oldoinyo Lengai, Kerimasi, Kwahera,
96 Hanang).

97 The Tarosero rocks have been dated to 2.4 - 1.9 Ma using K-Ar-dating (Macintyre *et al.*, 1974), with
98 the youngest ages being related to the rocks of the summit area. However, recent $^{40}\text{Ar}/^{39}\text{Ar}$ -dating
99 indicate 2.35 to 2.33 Ma for the peak region (Mana *et al.* (2015). The complex is a so-called lava cone
100 that extruded onto a nowadays northerly-dipping lava plateau of alkali basalts and trachyandesites of
101 which the volcanic center is unknown (Dawson, 2008). The complex is situated just west of the

102 Lembolos graben and mainly consists of trachyte lavas and minor trachytic tuffs, while primitive rocks
103 like basalts and more evolved peralkaline phonolites are also present (Fig. 1b).

104

105 **SAMPLE LOCALITIES**

106 In total, 73 samples deriving from 15 localities were examined in the course of this study. Of these, 28
107 samples were investigated in detail by means of electron microprobe and whole-rock analyses (Table
108 1). Samples were taken from the central part of the complex, including the cone itself, as well as from
109 the southern part, where more primitive to intermediate rocks (basalts, andesites, latites) occur. In a deep
110 gorge at the western flank of the complex, trachytic and phonolitic boulders were sampled. These derive
111 from eroded or almost inaccessible parts of the central area of the volcano and were transported by
112 ephemeral rivers.

113

114 **PETROGRAPHY**

115 The Tarosero rocks are classified according to the guidelines of the IUGS, Subcommittee on the
116 Systematics of Igneous Rocks (Le Maitre *et al.*, 2002). We distinguish the following six rock groups:

117 **Basalt**

118 Basalts contain up to 2 mm large euhedral olivine, typically rimmed by iddingsite (Fig. 2a). Chromian
119 spinel was found once as inclusion in olivine. Clinopyroxene occurs as up to several mm large, partly
120 rounded crystals (Fig. 2b) that sometimes reveal a slightly darker rim. The groundmass consists of
121 olivine, clinopyroxene, euhedral magnetite, rare ilmenite laths and tabular plagioclase.

122 **Andesite**

123 The andesite reveals a layered, tuff-like structure and contains up to several mm large macrocrysts of
124 euhedral to partly rounded amphibole, euhedral to subhedral clinopyroxene (Fig. 2c) and elongated
125 plagioclase crystals in a fine-grained groundmass (Fig. 2d). Magnetite occurs as phenocrysts (Fig. 2d)
126 and as inclusions in amphibole and clinopyroxene. Apatite is ubiquitous.

127 **Latite**

128 Latites contain several mm large and randomly arranged tabular plagioclase (Fig. 2e), euhedral to
129 subhedral and partly altered olivine, and subhedral clinopyroxene. Ilmenite and magnetite occur as
130 independent microphenocrysts, with the latter commonly surrounding ilmenite (Fig. 2f). The
131 groundmass mineralogy is similar to that of the phenocrysts with additional amphibole (Fig. 2f) and
132 apatite. The latter is occasionally replaced by monazite.

133 **Sodalite-bearing trachyte**

134 Sodalite-bearing trachytes (for the sake of simplicity in the following called “trachytes”) are
135 characterized by several mm large, randomly arranged plagioclase phenocrysts with a sieved core and a
136 rim of alkali feldspar (Fig. 2g). Euhedral and sometimes concentrically zoned clinopyroxene, euhedral
137 magnetite, subhedral amphibole (Fig. 2h) and rare ilmenite occur as phenocrysts and as inclusions in
138 plagioclase. Amphibole macrocrysts are surrounded by fine-grained clinopyroxene and magnetite (Fig.
139 2h). The groundmass mainly consists of alkali feldspar, and minor plagioclase, fluorite and subhedral to
140 interstitial sodalite. Carbonate was found in vugs within large plagioclase grains, apatite is ubiquitous.

141 **Peralkaline (sodalite-bearing) alkali feldspar trachyte and peralkaline phonolite**

142 Peralkaline (sodalite-bearing) alkali feldspar trachytes (in the following called “peralkaline trachytes”)
143 and peralkaline phonolites reveal a flow structure with oriented phenocrysts of tabular alkali feldspar.
144 Some of these rocks contain abundant dark brown amphibole macrocrysts (type I), bright green
145 clinopyroxene I and magnetite. Clinopyroxene I is typically enclosed by dark green clinopyroxene II
146 (Fig. 3a), while amphibole I is replaced by magnetite and bright brown amphibole II (Fig. 3b). The
147 amount of euhedral to subhedral sodalite, which commonly is altered to clay minerals and Fe-rich
148 hydroxides, strongly varies between the samples. Interstitial phases are alkali feldspar, clinopyroxene
149 II, aenigmatite and brownish amphibole IIIa (Fig. 3c), whereas late-stage interstitial blue amphibole IIIb
150 (Fig. 3d) only occurs in peralkaline phonolites with low contents of mafic phenocrysts. In peralkaline
151 phonolites, sodalite is commonly enclosed by amphibole IIIb and can contain inclusions of aenigmatite
152 (Fig. 3d). Some of the peralkaline phonolites contain interstitial eudialyte (Fig. 3e), which is, however,

153 frequently replaced by un-identified Zr-rich alteration phases. Further late-stage (probably
154 hydrothermal) phases are Mn-(hydr)oxides that mainly occur along cracks of alkali feldspar crystals and
155 rare REE-F-carbonates (bastnäsite; Fig. 3f). Apatite is rare. An agpaitic phonolite (TA-030) additionally
156 contains coarse-grained cumulates that mainly consist of alkali feldspar and dark brown amphibole I,
157 which is altered to bright brown amphibole II and magnetite when in direct contact to the peralkaline
158 phonolite (Fig. 3g). Agpaitic phonolite TA-021 contains a several mm large xenocrystic plagioclase
159 fragments with inclusions of clinopyroxene, magnetite and olivine.

160 **Peralkaline olivine-bearing alkali feldspar trachyte**

161 Peralkaline olivine-bearing alkali feldspar trachytes (in the following called “peralkaline olivine
162 trachytes”) are characterized by oriented alkali feldspar laths, subhedral magnetite, and olivine that is
163 commonly altered to iron (hydr)oxides (Fig. 3h) and also occurs as inclusions in alkali feldspar. The
164 groundmass mainly consists of small alkali feldspar laths, interstitial brown to dark blue amphibole and
165 aenigmatite (Fig. 3h). Apatite is very rare, and clinopyroxene is absent.

166

167 **METHODS**

168 **Whole-rock analyses**

169 For this study, major and trace elements of 25 rock samples were determined. Major element analyses
170 were done at the Department of Geosciences, University of Tübingen (Germany) with a wavelength
171 dispersive XRF (Bruker AXS S4 Pioneer; Rh-tube at 4kW) using 32 references materials (compiled in
172 Govindaraju, 1989). The rock samples were ground with an agate mill for 10 minutes. After drying at
173 105 °C, each 1.5 g of rock powder was mixed with 7.5 g MERCK spectromelt A12 (mixture of 66% Li-
174 tetraborate and 34% Li-metaborate), and melted at 1200°C to fused beads using an Oxiflux system (CBR
175 analytical service). Detection limits and analytical errors are in the 1-10 µg/g, and 0.1 % (relative) range,
176 respectively.

177 Trace elements were analysed by ICP-MS (X-series 2, Thermo Fisher Scientific) at the Institute for
178 Applied Geosciences, Karlsruhe Institute of Technology (Germany). Each 100 mg of powdered sample

179 was pre-oxidized (65 % HNO₃, subboiled) and digested in a mixture of 40 % HF (suprapur) and 65 %
180 HClO₄ (normatom) heated for 16 h at 120 °C in closed Teflon vessels to ensure complete silicate
181 decomposition. Afterwards, the acids were evaporated to incipient dryness and the residue was re-
182 dissolved in 65 % HNO₃ (subboiled) and evaporated again. This step was repeated three times for
183 purification purposes and the final residue was dissolved in 50 mL of ultrapure water. Due to the high
184 concentrations of Ba and light REE, mathematical corrections for Eu, Gd, Tb and Er were required:
185 ^{151/153}Eu was corrected for BaO, ¹⁵⁷Gd for PrO, ¹⁵⁸Gd for CeO and ¹⁵⁹Tb and ¹⁶⁷Er for NdO interferences.
186 The mean value of both isotopes are reported for Eu and Gd. The quality of the whole procedure was
187 ensured by adding six blanks, certified reference materials SY-2, SY-3 (Govindaraju, 1994) and CRM-
188 TMDW-A (High-Purity standards, Inc.) that were included in the digestion (accuracy: mainly ± 10 %).

189 Bulk halogen (F, Cl, Br) and S concentrations of 25 powdered whole-rock samples were determined by
190 Combustion Ion Chromatography (CIC) at the University of Tübingen. For this purpose, a 930 Compact
191 IC Flex chromatograph (Metrohm) with chemical suppression and a peristaltic pump for regeneration
192 (100 mmol/l H₂SO₄) connected to a combustion oven and an autosampler for solid samples (MMS 5000;
193 Analytik Jena), was used. Based on the frequent analysis of standard solutions (single element 1000
194 mg/l solutions; Roth), reference material GSN (granite), and replicate analyses for a subset of the
195 samples, the long-term reproducibility (1σ) for solid samples of the measurements is generally within a
196 10 % margin for F, Cl and S, while Br may vary by ~20 % because of its relatively low concentrations.
197 Effective detection limits are in the order of 10 µg/g for F and Cl, around 5 µg/g for S and 0.3 µg/g for
198 Br. See also Epp *et al.* (2019) and Epp *et al.* (2020) for a more detailed description of the method.

199 **Mineral analyses**

200 Mineral compositions were determined using a JEOL JXA 8900 electron microprobe in wavelength
201 dispersive mode at the Department of Geosciences, University of Tübingen (Germany). An acceleration
202 voltage of 15 kV was used for feldspar, sodalite, clinopyroxene, aenigmatite and amphibole, respectively
203 20 kV for olivine, oxides and eudialyte (see Supplementary Table 1 for a compilation of the WDS
204 configurations). Peak counting times for major elements were between 16 s and 30 s, and between 30
205 and 60 s for minor elements, while the background counting times were half as long. Both, synthetic

206 and natural standards were used for calibration. Peak overlap corrections involving Ti-V, V-Cr (oxides),
207 F-Fe (amphibole), Ba-Ti (feldspar), Hf-Ti and Hf-Sr (eudialyte) were implemented and the raw data
208 was converted using the implemented ZAF (for olivine, clinopyroxene, oxides, amphibole, aenigmatite)
209 and $\phi\rho z$ corrections (Armstrong, 1991). The beam current was set to 20 nA, except for sodalite and
210 eudialyte, which were analyzed with 5 nA. Sodium diffusion and destruction of the grains was
211 additionally avoided by widening the beam spot to 2 μm (feldspar, amphibole), respectively 5 μm
212 (sodalite). Eudialyte was also measured with a 5 μm beam spot (instead of the recommended 20 μm ;
213 e.g. Johnsen & Grice, 1999, Pfaff *et al.*, 2010) due to the small grain size. The standard deviation for
214 the major elements is typically $< 1\%$, but reaches up to $\sim 6\%$ for Fe in sodalite, and for Zr in eudialyte
215 as a result of the analytical conditions (low beam current) required to prevent the destruction of the
216 grains.

217 **Thermodynamic calculation of intrinsic parameters**

218 Several thermometers and (oxy)barometers, as well as thermodynamic calculations were used to
219 quantify the intrinsic parameters pressure (P), temperature (T), oxygen fugacity (f_{O_2}) and silica activity
220 (a_{SiO_2}) for suitable samples. Pressure and temperature were estimated using an amphibole
221 thermo(oxy)barometer (Ridolfi & Renzulli, 2012), and clinopyroxene-liquid barometers with their
222 pressure independent thermometers (Putirka *et al.*, 1996, Putirka *et al.*, 2003, Putirka, 2008, Neave &
223 Putirka, 2017). Whole-rock analyses were used as proxies for liquid compositions and special care was
224 taken to exclude samples that exhibit coarse-grained cumulates or xenoliths (see above). Temperature
225 was additionally estimated using ilmenite-magnetite equilibria Andersen & Lindsley (1985) applying
226 solid solution models of Stormer (1983), the QUILF software package Andersen *et al.* (1993), which is
227 based on coexisting olivine-Fe-Ti oxides \pm clinopyroxene, and feldspar solvus thermometry Fuhrman &
228 Lindsley (1988). Oxygen fugacity was calculated using the magnetite-melt oxybarometer of Arató &
229 Audétat (2017), which is based on the Fe and Ti partitioning between melt and mineral. Redox
230 conditions were also estimated by the ilmenite-magnetite oxybarometer, and both, f_{O_2} and a_{SiO_2} could be
231 determined using the QUILF software. Moreover, several redox and/or silica sensitive reactions
232 (reactions 1-4) were calculated using the Perple_X software package (Connolly (1990) with the

233 implemented database of Holland & Powell (1998). Aenigmatite-involving reaction (5) was executed
234 according to Marsh (1975).

235

236 (1): $2 \text{Fe}_3\text{O}_4$ (magnetite) + $3 \text{SiO}_2 = \text{O}_2 + 3 \text{Fe}_2\text{SiO}_4$ (fayalite)

237 (2): NaAlSiO_4 (nepheline) + $2 \text{SiO}_2 = \text{NaAlSi}_3\text{O}_8$ (albite)

238 (3): $\text{NaAlSi}_2\text{O}_6$ (jadeite) + $\text{SiO}_2 = \text{NaAlSi}_3\text{O}_8$ (albite)

239 (4): CaAlAlSiO_6 (Ca-Tschermak's pyroxene) + $\text{SiO}_2 = \text{CaAl}_2\text{Si}_2\text{O}_8$ (anorthite)

240 (5): Fe_3O_4 (magnetite) + $6 \text{SiO}_2 + 6 \text{NaFeSi}_2\text{O}_6$ (aegirine) + $3 \text{Fe}_2\text{TiO}_4$ (ulvöspinel) = $3 \text{Na}_2\text{Fe}_5\text{TiSi}_6\text{O}_{20}$
241 (aenigmatite) + 2O_2

242

243 All mineral equilibria calculations were corrected by the activities for endmember components of
244 olivine, clinopyroxene, Fe-Ti oxides and feldspar (Wood, 1979, Holland, 1990, Andersen *et al.*, 1993,
245 Holland & Powell, 2003). Aegirine and aenigmatite activities were calculated using mixing-on-site
246 models.

247 **Fractional crystallization modeling**

248 Fractional crystallization models were executed in the system $\text{SiO}_2\text{-TiO}_2\text{-Al}_2\text{O}_3\text{-Fe}_2\text{O}_3\text{-MnO-MgO-}$
249 $\text{CaO-Na}_2\text{O-K}_2\text{O-P}_2\text{O}_5$, assuming that phenocryst phases are primarily responsible for the fractionation
250 process. Whole-rock analyses were used as a proxy for the initial melt composition, while the target
251 composition was computed using the Least Square Fitting Method.

252

253 **RESULTS**

254 **Whole-rock data**

255 New whole-rock data of 25 samples, augmented by analyses from literature (Cooper, 1972, Mana *et al.*,
256 2015), represent the large compositional variation for the Tarosero rocks (Table 2). According to the

257 whole-rock chemistry using the TAS diagram (Fig. 4a) and further subdivision schemes (Le Bas *et al.*,
258 1986, Le Maitre *et al.*, 2002), the rocks can be mainly divided into basalts, basaltic trachyandesites
259 (shoshonite), trachyandesites (benmoreites) and trachytes. Trachytes are further subdivided into
260 comenditic and pantelleritic varieties (Fig. 4b; MacDonald, 1974). Note that the geochemical study of
261 Cooper (1972) did not mention eudialyte in the peralkaline trachytes/phonolites at all. This might be
262 related to the strong alteration of the grains, and/or to the very small size of the crystals which prevents
263 an easy recognition by transmitted light microscopy.

264 Concentrations of $\text{Na}_2\text{O}+\text{K}_2\text{O}$ and SiO_2 generally increase from primitive to evolved rocks, although
265 “fine-grained peralkaline trachytes” of Cooper (1972) show a rough negative correlation between
266 $\text{Na}_2\text{O}+\text{K}_2\text{O}$ and SiO_2 (Fig. 4a). The alkalinity index ($\text{A.I.}=[\text{Na}+\text{K}]/\text{Al}$) is lowest in basalts (~ 0.4) and
267 highest in eudialyte-bearing peralkaline phonolites (~ 1.2). A positive correlation exists between A.I. and
268 $\text{Na}_2\text{O}+\text{K}_2\text{O}$ (Fig. 4c), negative ones with CaO (Fig. 4d), TiO_2 (Fig. 4e) and MgO (Fig. 4f). The Al_2O_3
269 concentration rises in basalts towards the andesite, whereas trachytes and the peralkaline rocks show a
270 negative correlation with A.I. (Fig. 4g). Latites slightly deviate from this trend with an intermediate A.I.
271 and Al_2O_3 composition. A complex pattern is also illustrated by Fe_2O_3 (Fig. 5h) which decreases from
272 primitive to intermediate rock types, but increases towards agpaite peralkaline phonolites. The molar
273 Na/K ratio (Table 2) roughly decreases from the primitive rocks (molar Na/K=4.5-1.5) to the peralkaline
274 rock group, but is exceptionally low in the andesite (molar Na/K=1.5).

275 Compatible trace elements like Ni ($<440 \mu\text{g/g}$) and Cr ($<710 \mu\text{g/g}$) are only elevated within the basalts,
276 while incompatible elements such as REE (Fig. 5a) and HFSE (Fig. 5b & c) increase from the primitive
277 towards the peralkaline trachytes and peralkaline phonolites. Peralkaline trachyte TA-078 contains large
278 amounts of hydrothermal bastnäsite and has the highest REE concentrations, while peralkaline olivine
279 trachytes have rather low REE contents similar to the trachytes.

280 Primitive mantle-normalized (Palme & O'Neill, 2003) incompatible element patterns for basalts, the
281 andesite, latites and trachytes are very similar, except for the lack of negative P and Ti anomalies in
282 basalts (Fig. 6a), the peralkaline rocks are characterized by strongly negative Ba, Sr, P and Ti anomalies.
283 Primitive mantle-normalized REE abundances generally decrease from light REE (LREE) to heavy REE

284 (HREE), with increasing L_{aN}/L_{uN} ratios from basalts via latites-trachytes to peralkaline rocks. Negative
285 Eu anomalies are only present in the peralkaline rocks (Fig. 6b). Some of the peralkaline rocks, and most
286 pronounced the peralkaline olivine trachytes comprise strongly negative Ce anomalies, but have a less
287 prominent Eu anomaly (Fig. 6b).

288 Halogen concentrations are highly variable (Supplementary Table 2): Fluorine ranges between 680 and
289 2050 $\mu\text{g/g}$ and commonly shows a variation of more than 1000 $\mu\text{g/g}$ within a specific rock type (Fig.
290 7a). Chlorine (Fig. 7b) is typically below 250 $\mu\text{g/g}$ but much higher (up to ~ 2700 $\mu\text{g/g}$) for rocks
291 containing unaltered/relict sodalite grains (trachytes, peralkaline phonolites). Bromine mostly varies
292 between 0.7 and 3.7 $\mu\text{g/g}$, and SO_4 between 5 and 100 $\mu\text{g/g}$. The pattern for the F/Cl ratio (~ 1 -100) and
293 the Br/Cl ratio (~ 0.001 -0.1) are very similar (Fig. 7c & d), with samples containing fresh sodalite
294 illustrating the lowest values.

295 **Mineral data**

296 *Olivine*

297 Olivine compositions vary largely (Table 3; see Supplementary Table 2 for all mineral data), with
298 macrocrysts in basalts being Fa_{12-19} and groundmass olivines Fa_{39-42} . In latites, olivine is Fa_{41-60} , in
299 peralkaline olivine trachytes Fa_{66-85} and olivine in the plagioclase fragment of peralkaline phonolite TA-
300 021 has an intermediate composition (Fa_{48-50}). The tephroite component (0.2-6.2 mol%) increases with
301 the fayalite component (Fig. 8a). Nickel contents reach 0.008 apfu (~ 0.4 wt.% NiO) in basalts but are
302 otherwise mostly below the detection limit (0.01 wt.% NiO). Calcium contents range from 0.006 apfu
303 (~ 0.2 wt.% CaO) in basalts to 0.017 apfu (~ 0.5 wt.% CaO) in peralkaline olivine trachytes (Fig. 8b).

304 *Clinopyroxene*

305 Clinopyroxene evolves from diopside-hedenbergite_{ss} in primitive and intermediate rocks and reaches
306 about 80 mol% aegirine component in interstitial clinopyroxene of the peralkaline rocks (Table 4; Fig.
307 9a). Tschermak's components (CaXXSiO_6 ; X= Al, Ti, Fe^{3+}) reach 17 mol% in basalts and the andesite.
308 The Mn content (up to 0.04 apfu; 1.2 wt.% MnO) increases from the primitive to the evolved rock types,
309 but decreases with the aegirine content in interstitial clinopyroxenes of the peralkaline rocks (Fig. 9b).

310 Clinopyroxenes of the primitive rocks have the highest Ti (0.09 apfu; 3.3 wt.% TiO₂) and the lowest Zr
311 contents (generally below 0.001 apfu; 0.09 wt.% ZrO₂), the latter reaches up to 0.07 apfu (3.8 wt.%
312 ZrO₂) in peralkaline rocks (Figs. 9c & d).

313 *Oxides*

314 Spinel-group minerals represent magnetite-ulvöspinel solid solutions (Table 5; Fig. 10a). The Ti
315 concentrations decrease from the basalts (~75 mol% ulvöspinel) via the intermediate rock types (~65-
316 55 mol% ulvöspinel) towards the peralkaline trachytes/phonolites (~40-30 mol% ulvöspinel). Spinel-
317 group minerals in the peralkaline olivine trachytes, however, reveal rather high ulvöspinel components
318 between 70 and 60 mol%. Manganese (up to 0.08 apfu; 2.5 wt.% MnO) and Zn (0.011 apfu; 0.4 wt.%
319 ZnO) are positively correlated and are highest in peralkaline rocks (Fig. 10b). One basalt (TA-048)
320 additionally contains a Cr-spinel inclusion in olivine (Cr=1.1 apfu; Mg=0.5 apfu). Ilmenite has the
321 highest Ti content in the latites (Ilm₇₁₋₈₁Hem₁₁₋₁₆Prph₂₋₅Gk₃₋₁₃), followed by the basalts (Ilm₆₂₋₇₄Hem₁₄₋
322 ₂₇Prph₀₋₁Gk₉₋₁₄) and the trachytes (Ilm₆₄₋₆₈Hem₁₅Prph₃Gk₁₄₋₁₈).

323 *Aenigmatite*

324 Aenigmatite-group minerals are characterized by a low rhönite component and intermediate
325 compositions between the aenigmatite and the wilkinsonite endmember (Table 6; Fig. 11a). Mean Ti
326 contents increase from eudialyte-free peralkaline trachytes/phonolites (≈Aen₃₅₋₅₄Wlk₄₃₋₆₃Rhn₁₋₅) via
327 agpaitic phonolites (≈Aen₃₄₋₆₅Wlk₃₄₋₆₃Rhn₀₋₅) and peralkaline olivine trachytes (≈Aen₅₈₋₇₁Wlk₂₇₋₄₁Rhn₀₋
328 ₂). Eudialyte-free and eudialyte-bearing peralkaline trachytes/phonolites have similar Mg contents (0.1-
329 0.3 apfu; 0.05-1.3 wt.% MgO), while Zr contents (<0.4 apfu; 0.5 wt.% ZrO₂) of the former are on
330 average slightly higher (Fig. 11b). Peralkaline olivine trachytes have the lowest magnesium and
331 zirconium concentrations (≤0.1 apfu).

332 *Amphibole*

333 Amphibole compositions cover the whole spectrum from calcic to sodic species (Table 7). Amphibole
334 I (macrocrysts) reveals the highest Ca and lowest Si contents, while amphibole II (altered rim) and
335 especially amphibole III (interstitial) are more sodic and silicic (Fig. 12a; nomenclature according to

336 Hawthorne *et al.*, 2012). The concentration of F as well as X_{Mg} ($=Mg/[Mg+Fe^{2+}]$) increase from amp I
337 to amp II (Fig. 12b), while Ti decreases (Fig. 12c). Amphibole III has a large range for X_{Mg} (0-0.5),
338 which differs from amphibole IIIa ($X_{Mg}=0.4-0.5$) and amphibole IIIb ($X_{Mg}=0-0.3$). In peralkaline
339 phonolites, amph IIIb is low in Ti (<0.1 apfu) and F (1.2-1.6 apfu), while in amph IIIb Ti increases down
340 to $X_{Mg}=0.05$ and subsequently decreases for $X_{Mg}<0.05$. Fluorine in peralkaline olivine trachytes
341 negatively correlates with Ti, hence showing the reverse pattern with respect to X_{Mg} . The Zr
342 concentration is the highest for amp III with X_{Mg} between 0.15 and 0.55 (Fig. 12d), Cl is typically low
343 (≤ 0.01 apfu; 0.04 wt.%).

344 *Feldspar*

345 Plagioclase and alkali feldspar in the primitive to intermediate rocks vary in composition from
346 $An_{58}Ab_{40}Or_2$ over $An_{11}Ab_{78}Or_{11}$ to $An_2Ab_{58}Or_{40}$ (Table 8; Fig. 13a). Feldspars of the peralkaline rocks
347 are typically characterized by very low anorthite components ($An_1Ab_{70}Or_{29}$ to $An_0Ab_{63}Or_{37}$), while
348 higher Ca contents only occur in some feldspar cores of the peralkaline trachytes ($An_{12}Ab_{76}Or_{12}$), in the
349 cumulate of sample TA-030 ($An_{11}Ab_{4780}Or_{11}$) and the xenolith in sample TA-021 ($An_{21}Ab_{72}Or_7$).
350 Strontium is elevated (>0.01 apfu; 0.4 wt.% SrO) in plagioclase from andesite, trachytes and some
351 anorthite-rich alkali feldspars of the peralkaline trachytes. Iron reaches up to 0.08 apfu (2.2 wt.% Fe_2O_3)
352 in the peralkaline rocks, but is also elevated in the plagioclases of the basalts and the groundmass
353 plagioclases of the andesite (0.02-0.05 apfu; Fig. 13b). Barium is always below 0.01 apfu (0.5 wt.%
354 BaO).

355 *Sodalite*

356 Sodalite has almost Cl-endmember compositions (Table 9), with S typically being below the detection
357 limit (~ 0.05 wt.% SO_3). The analyzed sodalite grains contain minor K (mostly <0.15 apfu; Fig. 14a) and
358 Ca (<0.01 apfu; 0.06 wt.% CaO). Notable are rather high contents of Fe^{3+} (0.2-0.8 apfu) that correlate
359 negatively with Al (Fig. 14b), suggesting a coupled substitution between $Al+Fe^{3+}$ and Si (Fig. 14c).
360 Charge balancing may be maintained by the exchange mechanism of $Na^+(Al^{3+}, Fe^{3+}) = \square + Si^{4+}$.
361 Sodalites with elevated K, Si, and Fe^{3+} concentrations show low analytical totals down to 96 wt.%.

362 *Eudialyte*

363 Eudialyte (formula based on $\text{Si}+\text{Al}+\text{Zr}+\text{Ti}+\text{Hf}+\text{Nb}=29$ cations; Johnsen & Grice, 1999) reveals Na
364 contents of 11 to 14 apfu, and is characterized by Ca (3.6-5.1 apfu) and REE concentrations (0.3-0.8
365 apfu) which are typical of foid-bearing rocks (Table 10; Supplementary Figure 1a; Schilling *et al.*, 2011,
366 Estrade *et al.*, 2018). The contents of Fe (2.7-3.5 apfu) and Si (25.8-26.5 apfu) partly exceed the typical
367 maximum site occupation (Fe=3 apfu; Si=26 apfu). This might in part be related to the analytical
368 conditions (see above), although high Fe concentrations above 3 apfu for eudialyte from Tarosero were
369 also reported by Dawson (1997). We also consider the possibility that the formula calculation based on
370 the simplified assumption of $\text{Si}+\text{Al}+\text{Zr}+\text{Ti}+\text{Hf}+\text{Nb}=29$ cations may not be useful in all cases. Contrary
371 to the current literature data on eudialyte-group minerals, there is a positive correlation between Fe and
372 Mn (Supplementary Figure 1b). Zirconium varies between 2.0 and 2.9 apfu. Chlorine typically ranges
373 between 1.7 and 1.9 apfu, while strontium mostly is below the detection limit (<0.05 apfu; 0.16 wt.%
374 SrO).

375

376 **DISCUSSION**

377 **Crystallization conditions of the Tarosero rocks**

378 Intrinsic parameters like P, T, f_{O_2} , a_{SiO_2} , and $a_{\text{H}_2\text{O}}$ strongly affect the stability of mineral phases, and are
379 therefore of great importance to understand the evolution of magmatic systems. Based on these data and
380 their combination with whole-rock data and fractional crystallization calculations, we propose a possible
381 petrological evolution for the Tarosero Volcanic Complex.

382 *Crystallization pressure*

383 Pressure estimates for basalts were executed using the recently calibrated clinopyroxene-liquid
384 barometer of Neave & Putirka (2017) with the independent temperature estimate of Putirka (2008;
385 Equation 33). Temperatures range between 1140 and 1210 °C and pressures vary between 4 and 6 kbar,
386 corresponding to a crustal depth of 12 to 18 km. However, clinopyroxene-liquid models of Putirka *et al.*
387 (1996) and Putirka *et al.* (2003) imply much higher pressures of 11 ± 2 kbar, respectively ~ 25 -40 km

388 depth. This is consistent with the absence of plagioclase macrocrysts in the Tarosero basalts, since
389 experiments demonstrated that plagioclase typically does not occur as a near-liquidus phase above 10
390 kbar (Green & Ringwood, 1967). The pressure estimates conform with magmatic underplating at the
391 Moho, which is assumed to occur at depths < 35 km along the rift axis in southern Kenya (e.g. Prodehl
392 *et al.*, 1997). Pressure calculations for the intermediate andesite and the trachytes were determined by
393 applying the amphibole barometer of Ridolfi & Renzulli (2012). The calculated pressures of 7 kbar and
394 9 ± 1 kbar, respectively, demonstrate rather deep crystallization depths of 20-30 km, although the results
395 must be considered with caution since recent studies indicate that amphibole-related pressure estimates
396 are highly sensitive to the temperature and liquid composition (Erdmann *et al.*, 2014, Putirka, 2016).
397 However, the estimates are in good agreement with magma fractionation, ascent and storage at shallower
398 crustal levels compared to the more primitive basalts. Peccerillo *et al.* (2007) suppose that trachyte-
399 dominated volcanoes commonly form by high-pressure fractionation. Moreover, the fine-grained
400 reaction rim around the amphiboles and the sieve-textured plagioclase cores in trachytes (see above;
401 Figs. 3g & h) might reflect rapid decompression during ascent from a deep magma chamber to the
402 surface (e.g. Nelson & Montana, 1992, Rutherford & Devine, 2003, Buckley *et al.*, 2006). Peralkaline
403 magmas typically evolve in shallow-level magma plumbing systems of approximately 2-4 km depth
404 (e.g. Mahood, 1984, Novak & Mahood, 1986, Spandler & Morris, 2016, Marks & Markl, 2017). The
405 coarse-grained cumulate in peralkaline phonolite TA-030 (see above; Fig. 3g), consisting mainly of
406 large amphibole and feldspar crystals, is a further hint for a shallow magma chamber. In subsequent
407 calculations ($T, a_{\text{SiO}_2}, f_{\text{O}_2}$) on such a multi-level magmatic differentiation system, we assumed a pressure
408 of 11 kbar for the basalts, 9 kbar for the trachytes (and the latites), 7 kbar for the andesite and 1 kbar for
409 the peralkaline rocks. Calculations based on groundmass minerals were done at atmospheric conditions
410 (1 bar).

411 *Estimates for T, f_{O_2} and a_{SiO_2}*

412 Based on texturally coexisting olivine-clinopyroxene macrocryst pairs, near liquidus temperatures for
413 the basalts were determined to ~ 1000 °C, while clinopyroxene-liquid thermometers yield higher
414 temperatures (1290-1140 °C). The early groundmass assemblages indicate temperatures of 850-920 °C

415 (olivine-clinopyroxene-magnetite and ilmenite-magnetite, respectively). However, the latter two
416 estimates must be considered with caution since i) only one fresh ilmenite-magnetite pair was found (all
417 other groundmass magnetites of sample TA-048 are strongly altered to hematite) and ii) groundmass
418 olivines of TA-042 are completely altered to iddingsite, hence olivine-clinopyroxene-magnetite
419 equilibria could only be applied by combining mineral data of both samples (TA-048 and TA-042).
420 These results agree with minimum temperature estimates based on the feldspar solvus (850 °C; according
421 to Fuhrman & Lindsley, 1988). Computed redox conditions vary between $\Delta\text{FMQ} = -0.2$ and $\Delta\text{FMQ} +0.3$
422 (Fig. 15) with the relatively high redox conditions during groundmass formation being in good
423 accordance with the elevated Fe^{3+} -content in plagioclase (Fig. 13b). Silica activity calculations based on
424 olivine-clinopyroxene-magnetite equilibria (reactions 3+4) yield values of $a_{\text{SiO}_2} = 0.8$, whereas late
425 magmatic clinopyroxene-feldspar equilibria indicate a decrease of the silica activity to $a_{\text{SiO}_2} = 0.4 \pm 0.2$,
426 possibly due to the early clinopyroxene crystallization.

427 The early magmatic crystallization temperature for the andesite, the latites and the trachytes are
428 estimated to $\sim 1100\text{-}900$ °C using amphibole and ilmenite+magnetite \pm olivine assemblages. Groundmass
429 feldspars crystallized above 850 °C. Redox conditions for these rocks are typically rather reduced
430 ($\Delta\text{FMQ} = -1\text{-}0$; Fig. 15). However, the groundmass plagioclases of the andesite show similarly high Fe^{3+} -
431 concentrations compared to the basalts and may record increasing redox conditions during the late
432 magmatic stage (Fig. 13b). Silica activity can be constrained to $a_{\text{SiO}_2} = 0.4 \pm 0.2$ (reactions 3+4), except
433 for the andesite which shows a comparatively low silica activity of $a_{\text{SiO}_2} = 0.15$

434 The maximum temperature for the groundmass assemblage of the peralkaline rock groups can be
435 assumed to be around 750 °C ± 50 based on experimentally synthesized aenigmatite and arfvedsonite
436 (Ernst, 1962, Bailey, 1969, Thompson & Chisholm, 1969, Scaillet & Macdonald, 2001, Giehl *et al.*,
437 2013, Giehl *et al.*, 2014), consistent with a feldspar solvus ≥ 700 °C. Redox conditions during
438 crystallization of the peralkaline trachytes and peralkaline phonolites was roughly estimated by an
439 equilibrium reaction including magnetite, clinopyroxene and aenigmatite (reaction 5). Since oxides
440 typically do not coexist with aenigmatite (Nicholls & Carmichael, 1969), this reaction reflects the redox
441 conditions that prevailed during termination of magnetite and onset of aenigmatite crystallization. The

442 high uncertainty of the exact mineral compositions during this transition results in a large possible redox
443 range of $\Delta\text{FMQ}=-5$ to $\Delta\text{FMQ}=0$ (Fig. 15) and a high uncertainty for the silica activity ($a_{\text{SiO}_2}=0.2-0.8$).
444 Oxygen fugacity estimates for the peralkaline olivine trachytes ($\Delta\text{FMQ}=-2.6-0$) were computed using
445 coexisting olivine, magnetite, and a silica activity >0.3 due to the absence of nepheline but the presence
446 of alkali feldspar (reactions 1+2).

447 **Source region for the rocks from Tarosero**

448 The high concentrations of incompatible elements in peralkaline rocks are commonly related to a
449 metasomatically enriched mantle source (Martin *et al.*, 1994, Upton *et al.*, 2003, Markl *et al.*, 2010)
450 which is also assumed for Tarosero based on trace element fractionation models (Mana *et al.*, 2015).
451 The low Cs and Rb abundances and the negative K anomaly in the primitive rocks from Tarosero suggest
452 the presence of a K-bearing mineral phase in the mantle source (Fig. 6a). Since the primitive mantle-
453 normalized Ba/Ce ratio of ≥ 1 illustrates a residual phase that only incorporates minor amounts of Ba,
454 we suggest that amphibole rather than phlogopite is the dominant potassium phase in the mantle (Mana
455 *et al.*, 2015). However, contributions of other residual mineral phases like olivine, pyroxene or garnet
456 increase the Ba/Ce ratio, and hence, might be balanced by the presence of phlogopite (Halliday *et al.*,
457 1995). Radiogenic Sr-Nd-Pb isotope values indicate the involvement of several mantle components,
458 with the least evolved samples trending towards the isotope composition of the Kenya Plume (Aulbach
459 *et al.*, 2011), while more evolved rocks might be contaminated by crustal components (Mana *et al.*,
460 2015).

461 **Genetic relationship between the rock types at Tarosero**

462 Continuous trends in mineral compositions (e.g. clinopyroxene, magnetite, amphibole) suggest a genetic
463 link between the investigated rock types, as do the very similar ages for primitive basalts and the evolved
464 rocks of the volcano (Macintyre *et al.*, 1974, Mana *et al.*, 2015). Besides basanitic and nephelinitic
465 melts, alkali basaltic melts are candidates for the source of peralkaline rocks (e.g. Kramm & Kogarko,
466 1994, Schilling *et al.*, 2011, Andersen *et al.*, 2014, Estrade *et al.*, 2014, Spandler & Morris, 2016). In
467 the case of Tarosero, negative Eu anomalies in the evolved rocks (Fig. 6b) indicate extensive plagioclase
468 fractionation, which is a common characteristic of many peralkaline rocks (e.g. Sørensen, 1992, Ridolfi

469 *et al.*, 2006) and is typical of their derivation from basaltic, rather than from basanitic or nephelinitic
470 magmas (Marks *et al.*, 2011). Similarly, the low Sr concentrations in eudialyte (Table 10) and the
471 negative Sr anomaly in the peralkaline suite (Fig. 6a) supports this assumption (e.g. Kramm & Kogarko,
472 1994, Schilling *et al.*, 2011).

473 *Evolution from the primitive to intermediate compositions*

474 Based on the arguments presented above, we computed fractional crystallization models. Due to the
475 exceptionally low Na/K ratio (Table 2) and the low silica activity of the andesite, we excluded this rock
476 type from further modelling and suggest that it is derived from a different magma source. In the absence
477 of a better suited primitive rock type, we first tested if the basalts from Tarosero represent suitable
478 parental magmas for the latites and the more evolved rocks.

479 Fractional crystallization based on the phenocryst phases olivine and clinopyroxene (path 1a in Fig. 16)
480 does not result in an evolutionary path from the basaltic composition towards the intermediate latites
481 (note that no clinopyroxene fractionation is required for the best fit; see Supplementary Table 3) nor
482 does the addition of titanomagnetite (path 1b) or titanomagnetite + plagioclase fractionation (path 1c).
483 While the fractionation of titanomagnetite could be suggested by negative Ti anomalies (Fig. 6a), the
484 lack of a negative Eu and Sr anomaly in latites argues against extensive plagioclase fractionation. We
485 conclude that the investigated basalts do not represent a parental magma for the latites. Unfortunately,
486 no other primitive rock type was sampled in the field which could represent a suitable parental magma
487 composition. Since extensive plagioclase fractionation can be ruled out for the rather high alkalinity
488 index A.I. of the latites (A.I.=0.74-0.84), we assume that these rocks derive from a mantle source
489 involving partial melting of phlogopite (A.I. \approx 1). However, we emphasize that both, the basalts and the
490 latites might also represent a tectonic window of the underlying plateau lavas for which the genetic
491 relationship to the more evolved rock types at Tarosero is still enigmatic (Dawson, 2008).

492 *Evolution models from intermediate to the evolved (peralkaline) rocks*

493 The evolution from latites towards trachytes can be modelled by fractionation of minor amounts (<10
494 %) of clinopyroxene, magnetite and olivine. However, the slightly elevated alkalinity index of trachytes

495 compared to latites additionally requires minor plagioclase fractionation (path 2 in Fig. 16). The lack of
496 a Eu anomaly in the whole-rock analyses could in part be explained by fractional crystallization of
497 apatite (Watson & Green, 1981), as indicated by the negative P anomaly (Fig. 6a). Absence of intense
498 plagioclase fractionation is also demonstrated by the increasing Sr concentration in plagioclase (Fig.
499 13b) and whole-rock analyses (Fig. 6a) from latites towards trachytes. Fractionation of the mafic
500 minerals is consistent with the steeper REE pattern compared to the more primitive rocks (e.g. White,
501 2013). The presence of ilmenite in both rock types might be a further hint for a genetic link. Late,
502 interstitial amphibole in latites becomes an early precipitating mineral phase in the trachytes,
503 presumably due to the increase of the water concentration of the melt because of ongoing fractionation
504 of water-free minerals.

505 The presence of amphibole in trachytes and the more evolved peralkaline trachytes/ phonolites could be
506 a hint for a close genetic relationship between them. The composition of the latter can be reached by
507 fractionation of large amounts (>20 %) of plagioclase, minor magnetite, and traces (< 1 %) of ilmenite
508 and apatite (path 3), in addition to minor contamination and assimilation with SiO₂-rich rocks (for
509 simplicity modelled as quartzite). Contamination with crustal material for the evolved rocks at Tarosero
510 has already been proposed by Mana (2013) and Mana *et al.* (2015) based on the trace element modelling
511 and Sr-Nd-Pb isotope data. As soon as the melt becomes peralkaline, plagioclase crystallization is
512 suppressed (Scaillet & Macdonald, 2001). This is in agreement with the absence of plagioclase in the
513 peralkaline trachytes and phonolites. The agpaitic compositions of the peralkaline phonolites can be
514 reached by further fractionation of alkali feldspar, minor amphibole and traces of magnetite and apatite
515 (path 4).

516 Peralkaline olivine trachytes are mineralogically distinct as they lack clinopyroxene and calcic
517 amphibole (relics) but contain olivine instead (see above). Based on the mineralogy (olivine, Fe-Ti-
518 oxides; interstitial amphibole) and the mineral chemistry (Fig. 9, Fig. 12), these rocks are closely related
519 to latites. We therefore suggest that they derive from latites by fractionation of plagioclase, minor
520 clinopyroxene, olivine, magnetite and apatite, accompanied by minor crustal contamination (path 5).

521

522 **Hydrothermal overprint of the peralkaline rocks**

523 Several peralkaline rocks at Tarosero are characterized by negative Ce anomalies in their primitive
524 mantle-normalized REE patterns (Fig. 17a). The low solubility of Ce^{4+} (Bouchaud *et al.*, 2012) in
525 aqueous fluids compared to REE in their trivalent oxidation state directly affects the REE distribution
526 in hydrothermally altered samples. Leaching of REE by an aqueous fluid results in a positive Ce
527 anomaly in the peralkaline rocks, while precipitation of REE-bearing minerals from the same fluid has
528 the opposite effect. Indeed, peralkaline trachyte TA-078 contains the highest REE concentrations of all
529 samples (Fig. 5) due to the hydrothermal formation of bastnäsité (Fig. 3f), consistent with a strongly
530 negative Ce anomaly. Since Eu^{2+} is more soluble than Eu^{3+} at temperatures $> 250\text{ }^{\circ}\text{C}$ (Sverjensky, 1984,
531 Bilal, 1991), the (high temperature) hydrothermal overprint should have the opposite effect on Eu
532 abundances. However, the strongly negative Eu anomaly in the peralkaline rocks, caused by the intense
533 fractionation of plagioclase, generally may hide this effect. Moreover, peralkaline rocks with high initial
534 REE concentrations are somewhat buffered against distinctive REE anomalies (Fig. 17b). Hence,
535 peralkaline olivine trachytes with their comparatively low REE concentrations show the strongest
536 negative Ce anomalies and are also characterized by a comparably small Eu anomaly (Fig. 17a; Fig. 6b).

537 **Constraints for the formation of extrusive agpaitic rocks**

538 Agpaitic rocks are typically related to intrusive complexes, and the occurrence of extrusive agpaitic
539 rocks is restricted to rather few localities (e.g. at Oktyabr'sky, Tarosero and Kontozero; Krivdik &
540 Tkachuk, 1988, Dawson, 1997, Petrovsky *et al.*, 2012). The rarity of volcanic equivalents might, in part,
541 arise from erosion, alteration, or overlooking of small-grained agpaitic minerals. However, the major
542 prerequisite for the formation of agpaitic rocks is the retention of the volatile halogens in the magma
543 during fractionation (e.g. Marks & Markl, 2001, Marks & Markl, 2015). This requires special conditions
544 which are discussed in the following section.

545 *General constraints for the formation of agpaitic rocks*

546 Most agpaitic rocks are plutonic and hence, indicate a magmatic system that prevents the easy escape
547 of volatile phases (Sørensen, 1992). Redox conditions below the FMQ buffer are a further common

548 characteristic for agpaitic rocks (Marks & Markl, 2017 and references therein). This is in agreement
549 with the fact that low redox conditions suppress the early crystallization of magnetite, thereby increasing
550 the Fe concentrations of the melt. High Fe concentrations in turn enhances the solubility of halogens
551 like Cl (Metrich & Rutherford, 1992, Giehl *et al.*, 2013), which is required for the formation of agpaitic
552 minerals. Besides the redox conditions, the water activity is the crucial parameter in the evolution of
553 agpaitic complexes since a low $a_{\text{H}_2\text{O}}$ typically prevents expulsion of water-rich fluids. This prevention
554 results in an enrichment of typically water-soluble elements like alkali metals and halogens in the melt
555 (e.g. Kogarko, 1987, Markl *et al.*, 2001, Liebscher & Heinrich, 2007, Marks & Markl, 2015), which
556 increase the solubility for REE and HFSE (Watson, 1979, Keppler, 1993, Linnen & Keppler, 2002) and
557 support formation of HFSE-enriched minerals.

558 The necessary conditions for the formation of agpaitic minerals can be reached during orthomagmatic,
559 late-magmatic and hydrothermal stages (Marks & Markl, 2017). In the agpaitic phonolite lavas from
560 Oktyabr'sky (Krivdik & Tkachuk, 1988) and Kontozero (Petrovsky *et al.*, 2012), eudialyte occurs as
561 (micro)phenocrysts, which indicates melt saturation in HFSE and chlorine at early magmatic stages prior
562 to the lava extrusion. In contrast, peralkaline lavas at Tarosero may contain interstitial eudialyte,
563 illustrating a process that retains halogens in the late-magmatic melt residue.

564 *Crystallization conditions and implications for the formation of the peralkaline trachytes/phonolites*

565 The oxygen fugacity during the formation of the peralkaline trachytes/phonolites was estimated to
566 $\Delta\text{FMQ}=-5-0$, seemingly low enough to suppress extensive magnetite fractionation, which caused Fe-
567 enrichment during evolution towards the most evolved peralkaline phonolites (Figs. 4b & c). The
568 increasing Fe concentrations enhance the Cl solubility of the melt, as demonstrated by the increasing
569 amount of orthomagmatic to late-magmatic sodalite (Fig. 3d). Experimental work of Giehl *et al.* (2013)
570 illustrated that besides low redox conditions, a low water activity is a further prerequisite to suppress
571 magnetite crystallization. We suggest that the water activity during evolution from the trachytes to the
572 peralkaline magma was kept low due to the near liquidus crystallization of OH-rich amphibole. The
573 relatively dry late-magmatic conditions prevented fluid exsolution and the escape of volatile elements
574 such as alkali metals and halogens. The enrichment of these elements during magma evolution increased

575 the solubility for REE and HFSE, and supported the late-stage formation of interstitial eudialyte. Due to
576 the onset of eudialyte crystallization, the most evolved clinopyroxenes (Fig. 9d) and amphiboles (Fig.
577 12d), as well as aenigmatite in the eudialyte-bearing peralkaline phonolites (Fig. 11b) reveal low Zr
578 concentrations (Markl *et al.*, 2001). However, there are several peralkaline phonolites which do not
579 contain eudialyte but still have high concentrations of HFSE (and REE). We suggest that late-stage
580 clinopyroxene with up to ~3 wt. % ZrO₂ (Table 4) rather than altered eudialyte dominates the HFSE
581 budget in these rocks (q.v. Andersen *et al.*, 2012).

582 Notable, the intermediate trachytes show similarly high F concentrations and even higher Cl
583 concentrations (Figs. 7a & b) than most of the agpaitic phonolites. Rather high F concentrations in the
584 trachytes might be related to the dry late-magmatic conditions which impeded the loss of F by exsolving
585 fluids and resulted in the crystallization of fluorite. Increasing F concentrations during magma evolution
586 towards the agpaitic phonolites might have been prevented by fractionation of amphibole with up to 0.5
587 apfu F (Fig. 12b), although the relatively large scatter in F concentrations of both rock types may suggest
588 variable F loss to the atmosphere. Moreover, the variable F concentrations are in part related to a
589 hydrothermal overprint as demonstrated by sample TA-078 which contains abundant REE-F-carbonates
590 (bastnäsite; Fig. 3f; Fig. 7a). The relatively low Cl contents in the peralkaline rocks may also result from
591 post-magmatic alteration processes. While interstitial sodalite in the trachytes is pristine, sodalite in the
592 peralkaline trachytes/phonolites is strongly altered and replaced by (Cl-free) Fe-hydroxides and clay
593 minerals (note that the beginning of sodalite alteration is marked by constant Cl but, elevated K, Si, and
594 Fe³⁺ concentrations, and low analytical totals which may demonstrate microscale porosity; Fig. 14). We
595 suppose that the initial Cl concentration of the peralkaline magma was much higher than that of the
596 trachytes, hence evolving towards REE- and HFSE-rich compositions that finally crystallized interstitial
597 eudialyte.

598 *Crystallization conditions and implications for the formation of the peralkaline olivine trachytes*

599 Peralkaline olivine trachytes at Tarosero reveal much lower REE and HFSE concentrations compared
600 to the peralkaline trachytes/phonolites (Fig. 5) and do not contain agpaitic minerals. A discrimination of
601 the two peralkaline rock series based on the oxygen fugacity is unfeasible because of the high uncertainty

602 of the redox calculations. However, we suggest that the peralkaline olivine trachytes formed under high
603 water activities, which are indicated by the lack of clinopyroxene (Bailey, 1969, Scaillet & Macdonald,
604 2001, Andersen & Sørensen, 2005), and the higher OH-contents in the interstitial amphiboles compared
605 to those of the peralkaline trachytes/phonolites (Fig. 12b). The absence of sodalite is further evidence
606 for a high water activity since preliminary work of Giehl *et al.* (2014) demonstrated sodalite stability
607 only for peralkaline melts with low H₂O concentrations. An increasing water activity during magma
608 evolution results from the fractionation of olivine instead of OH-rich amphibole. Fractionation under
609 relatively low crustal pressures, and hence low fluid pressures, may have prevented the change from
610 olivine to amphibole as the dominant phenocryst phase during evolution from the latites towards the
611 trachytes (e.g. Ernst, 1962). We therefore suppose that the formation of the peralkaline olivine trachytes
612 was accompanied by significant amounts of fluid exsolutions. The consequent loss of volatiles phases
613 resulted in a peralkaline melt with low solubilities of HFSE and REE. Furthermore, exsolving halogen-
614 rich aqueous fluids have the potential to transport significant amounts of HFSE which additionally leads
615 to a decrease of these elements in the remaining melt (Migdisov *et al.*, 2009, Migdisov *et al.*, 2011,
616 Timofeev *et al.*, 2015).

617 The high water activity presumably prevented the crystallization of (Ti-rich) aenigmatite during the
618 beginning of interstitial amphibole formation. Hence, in contrast to the eudialyte-bearing peralkaline
619 phonolites, the Ti content of amphibole increased during evolution of the peralkaline olivine trachytes
620 (Fig. 12c). Contrary to most peralkaline trachytes/phonolites, the lack of fractionation of F-bearing
621 amphibole phenocrysts resulted in rather high F concentrations in the peralkaline olivine trachytes (Fig.
622 7a). Late-stage aenigmatite formation might be correlated with a sudden lowering of the water activity
623 in the melt residue due to the expulsion of fluids, as indicated by the changing OH/F and Ti pattern in
624 amphibole at X_{Mg}=0.5 (Fig. 12c). We speculate that during this process, most of the remaining REE,
625 HFSE and chlorine could escape from the melt residue, hence preventing the crystallization of eudialyte.

626

627

628 SUMMARY AND CONCLUSION

629 The Tarosero volcano covers a broad spectrum from primitive to evolved rock types that formed by
630 partial melting of an amphibole- and phlogopite-bearing mantle combined with multi-level
631 differentiation at depths ≤ 35 km. All rock types formed under relatively reduced conditions (\leq FMQ
632 buffer), which is typical of peralkaline complexes associated with agpaitic rocks.

633 Basalts represent the most primitive rocks at Tarosero but fractional crystallization models cannot cross
634 the gap towards the latites, indicating that the basalts do not reflect their parental magma source. The
635 high alkalinity index A.I. of the latites presumably illustrates a derivation from a mantle source involving
636 partial melting of phlogopite. However, starting from the intermediate latites, fractional crystallization
637 of mainly plagioclase, and variable amounts of clinopyroxene, olivine, magnetite and ilmenite,
638 accompanied by minor crustal contamination can reproduce the observed composition of peralkaline
639 rocks. The most evolved peralkaline phonolites require additional alkali feldspar and minor amphibole
640 fractionation. They are characterized by interstitial eudialyte, which indicates that halogens were (partly)
641 prevented from escaping to the atmosphere. Generally, agpaitic rocks are plutonic to subvolcanic rocks,
642 hence illustrating a magmatic system that prevents the easy loss of volatile phases. In case of the
643 extrusive rocks at Tarosero, fluid exsolution and loss of water-soluble halogens was impeded by a low
644 water activity in the late-magmatic conditions caused by the early crystallization of OH-bearing
645 amphibole. In contrast, peralkaline olivine trachytes are characterized by much lower HFSE and REE
646 concentrations and the absence of agpaitic minerals. We suggest that melt evolution towards this rock
647 type occurred at lower pressure, hence crystallizing fayalite instead of amphibole phenocrysts. The lack
648 of an early water-sink led to an increase of the water activity and the expulsion of halogen-bearing fluids.
649 Finally, this prevented the crystallization of agpaitic assemblages in these rocks.

650

651

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658

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662

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Table 1: List of investigated samples. Locality numbers as indicated in Fig. 1b.

Locality #	Rock type	Sample #	Coordinates (WGS84)		Mineral assemblage											EPMA data	WR data
			Latitude (°N)	Longitude (°E)	Ol	Cpx	Amp	Ox	Aen	Fsp	Sdl	Eud					
1	Peralkaline phonolite (agpaitic)	TA-008	-3.192030	36.359916		x	x	x	x	x	x	x	x	x	x	x	x
2	Peralkaline olivine trachyte	TA-005	-3.206808	36.353964	x		x	x	x	x							x
2	Peralkaline olivine trachyte	TA-006	-3.201757	36.354662	x		x	x	x	x							x
2	Peralkaline olivine trachyte	TA-007	-3.197472	36.356970	x		x	x	x	x							x
6	Trachyte	TA-116	-3.224889	36.325347		x	x	x	x	x							x
6	Trachyte	TA-118	-3.224892	36.325351		x	x	x	x	x							x
7	Peralkaline trachyte	TA-054	-3.228092	36.313845		x	x	x	x	x							x
7	Peralkaline trachyte	TA-056	-3.228055	36.313871		x	x	x	x	x							x
7	Peralkaline phonolite	TA-059	-3.228727	36.314917		x	x	x	x	x							x
7	Peralkaline phonolite	TA-065	-3.228132	36.316177		x	x	x	x	x							x
7	Peralkaline phonolite (agpaitic)	TA-070	-3.230712	36.317713		x	x	x	x	x							x
7	Peralkaline phonolite (agpaitic)	TA-072	-3.230712	36.317713		x	x	x	x	x							x
7	Peralkaline phonolite	TA-073	-3.230712	36.317713		x	x	x	x	x							x
9	Peralkaline trachyte	TA-078	-3.254544	36.317459		x	x	x	x	x							x
9	Peralkaline phonolite	TA-079	-3.254544	36.317459		x	x	x	x	x							x
9	Peralkaline phonolite	TA-080	-3.254887	36.316688		x	x	x	x	x							x
10	Latite	TA-045	-3.293488	36.301786	x		x	x	x	x							x
11	Basalt	TA-042	-3.306456	36.301439	x		x	x	x	x							x
11	Andesite	TA-044	-3.303982	36.301464		x	x	x	x	x							x
12	Basalt	TA-048	-3.313556	36.291862	x		x	x	x	x							x
12	Basalt	TA-049	-3.313556	36.291862	x		x	x	x	x							x
12	Latite	TA-046	-3.313651	36.293727	x		x	x	x	x							x
15	Trachyte	TA-026	-3.227314	36.234362		x	x	x	x	x							x
15	Peralkaline phonolite (agpaitic)	TA-021	-3.227316	36.235095	x		x	x	x	x							x
15	Peralkaline phonolite (agpaitic)	TA-022	-3.227497	36.235158		x	x	x	x	x							x
15	Peralkaline phonolite (agpaitic)	TA-030	-3.227910	36.233296		x	x	x	x	x							x
15	Peralkaline phonolite	TA-034	-3.227886	36.231895		x	x	x	x	x							x
15	Peralkaline phonolite	TA-035	-3.227886	36.231895		x	x	x	x	x							x

Abbreviations: Ol=Olivine; Cpx=Clinopyroxene; Amp=Amphibole; Ox=Oxides; Fsp=Feldspar; Sdl=Sodalite; Aen=Aenigmatite; Eud=Eudialyte; EPMA=electron microprobe analysis; WR=whole-rock

Table 3: Selected EPMA analyses for olivine from the Tarosero Volcanic Complex.

Rock type	Basalt		Latite		Peralkaline phonolite (agpaitic)		Peralkaline olivine trachyte	
	TA-042_1 macrocryst	TA-048 groundmass	TA-045 macrocryst	TA-046 groundmass	TA-021 xenolith	TA-005 macrocryst	TA-005 macrocryst	
wt. %								
SiO ₂	40.60	35.28	34.75	33.62	34.34	30.64	32.02	
TiO ₂	bdl	0.07	0.03	0.08	0.02	0.02	0.04	
Al ₂ O ₃	0.03	0.02	bdl	0.02	0.02	0.00	bdl	
FeO	15.07	36.53	39.24	45.48	41.11	58.82	53.69	
MnO	0.20	0.54	1.37	1.80	1.73	4.25	3.37	
MgO	44.58	27.73	25.06	19.24	23.02	6.37	11.74	
NiO	0.31	0.07	bdl	0.02	bdl	bdl	bdl	
CaO	0.25	0.27	0.30	0.39	0.29	0.46	0.36	
Total	101.04	100.51	100.76	100.65	100.52	100.58	101.23	
<i>Formula based on 4 oxygens</i>								
Si	1.01	0.98	0.98	0.99	0.99	0.99	0.99	
Ti	bdl	0.00	0.00	0.00	0.00	0.00	0.00	
Al	0.00	0.00	bdl	0.00	0.00	0.00	bdl	
Fe	0.31	0.85	0.93	1.12	0.99	1.59	1.38	
Mn	0.00	0.01	0.03	0.04	0.04	0.12	0.09	
Mg	1.65	1.15	1.06	0.84	0.99	0.31	0.54	
Ni	0.01	0.00	bdl	0.00	bdl	bdl	bdl	
Ca	0.01	0.01	0.01	0.01	0.01	0.02	0.01	
Sum cations	2.99	3.01	3.01	3.01	3.01	3.01	3.01	

bdl: below detection limit

Table 4: Selected EPMA analyses for clinopyroxene from the Tarasero Volcanic Complex.

Rock type Thin section	Basalt		Andesite		Latite		Trachyte		Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)		
	TA-042_1 large	TA-048 Small	TA-044 large	TA-046 large	TA-046 Small	TA-116 large	TA-118 Small	TA-073 large	TA-054 interstitial	TA-072 large	TA-072 interstitial	TA-050 cumulate	TA-021 xenolith
wt. %													
SiO ₂	53.84	49.31	48.11	51.63	51.56	52.39	49.61	52.34	50.48	52.27	50.87	52.35	52.09
TiO ₂	0.51	2.03	2.12	0.74	1.05	1.01	1.77	0.32	0.24	0.40	0.27	0.36	0.61
ZrO ₂	bdl	bdl	0.03	bdl	0.03	0.04	0.82	0.05	2.55	0.05	3.27	0.04	0.03
Al ₂ O ₃	1.99	3.89	5.65	1.75	1.57	2.30	1.49	0.51	0.46	0.71	0.33	0.59	1.19
Cr ₂ O ₃	0.53	0.13	0.04	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Fe ₂ O ₃ *	0.02	1.97	3.29	1.57	0.90	0.00	3.78	1.64	16.57	1.70	19.28	1.65	3.08
FeO*	5.34	9.31	6.39	8.87	10.20	9.48	13.69	13.50	10.45	13.47	7.28	15.22	9.00
MnO	0.13	0.24	0.33	0.54	0.59	0.54	0.98	1.04	0.94	10.30	0.67	1.20	0.72
MgO	17.92	13.30	11.28	14.13	13.21	12.42	6.11	9.83	0.95	10.30	0.95	8.83	13.64
CaO	20.02	19.68	22.27	20.08	20.30	21.12	19.42	20.22	8.77	19.58	6.60	19.64	20.24
Na ₂ O	0.37	0.49	0.88	0.45	0.47	0.89	2.32	1.06	8.13	1.07	9.68	1.21	0.62
K ₂ O	0.02	bdl	bdl	bdl	0.02	0.02	0.07	bdl	0.02	bdl	bdl	bdl	bdl
F	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Cl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Total	100.67	100.36	100.39	99.75	99.91	100.21	100.03	100.51	99.56	100.58	99.20	101.08	101.22
<i>Formula based on 4 cations and 6 oxygens</i>													
Si	1.95	1.85	1.80	1.93	1.94	1.96	1.93	1.99	2.00	1.99	2.01	2.00	1.94
Ti	0.01	0.06	0.06	0.02	0.03	0.03	0.05	0.01	0.01	0.01	0.01	0.01	0.02
Zr	bdl	bdl	0.00	bdl	0.00	0.00	0.02	0.00	0.05	0.00	0.06	0.00	0.00
Al	0.08	0.17	0.25	0.08	0.07	0.10	0.07	0.02	0.02	0.03	0.02	0.03	0.05
Cr	0.02	0.00	0.00	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Fe ³⁺	0.00	0.06	0.09	0.04	0.03	0.00	0.11	0.05	0.49	0.05	0.57	0.05	0.09
Fe ²⁺	0.16	0.29	0.20	0.28	0.32	0.30	0.45	0.43	0.35	0.43	0.24	0.49	0.28
Mn	0.00	0.01	0.01	0.02	0.02	0.02	0.03	0.03	0.03	0.03	0.02	0.04	0.02
Mg	0.97	0.74	0.63	0.79	0.74	0.69	0.35	0.56	0.06	0.58	0.06	0.50	0.76
Ca	0.78	0.79	0.89	0.81	0.82	0.84	0.81	0.83	0.37	0.80	0.28	0.80	0.81
Na	0.03	0.04	0.06	0.03	0.03	0.06	0.17	0.08	0.62	0.08	0.74	0.09	0.04
K	0.00	0.00	bdl	bdl	0.00	0.00	0.00	bdl	0.00	bdl	bdl	bdl	bdl
Total cations	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00
F	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Cl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Total anions	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00
Mg#	0.85	0.71	0.75	0.73	0.69	0.69	0.43	0.55	0.13	0.56	0.18	0.49	0.71

bdl: below detection limit; *Fe2O3-FeO ratio was calculated.

Table 5: Selected EPMA analyses for magnetite and ilmenite from the Tarosero Volcanic Complex.

Rock type Thin section Mineral	Basalt		Andesite		Latite		Trachyte		Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)		Peralkaline olivine tract	
	TA-048 Cr-Spinel	TA-042_1 Magnetite	TA-048 Ilmenite	TA-044 Magnetite	TA-045 Magnetite	TA-046 Ilmenite	TA-118 Magnetite	TA-116 Ilmenite	TA-054 Magnetite	TA-73 Magnetite	TA-008_2 Magnetite	TA-030 Magnetite cumulate	TA-021 Magnetite xenolith	TA-006 Magnetite
wt. %														
SiO ₂	0.08	0.08	0.05	0.09	0.05	bdl	0.07	bdl	0.06	0.08	0.13	0.07	0.05	0.09
TiO ₂	2.01	25.60	45.04	22.20	22.97	49.47	20.49	48.66	16.14	14.91	14.74	20.14	21.46	21.07
Al ₂ O ₃	13.56	0.92	0.14	3.56	2.34	0.08	1.83	0.15	0.35	0.37	0.08	0.47	1.59	bdl
V ₂ O ₅	0.20	0.32	0.19	0.09	0.18	bdl	0.07	0.03	0.10	0.05	0.03	0.13	0.16	bdl
Nb ₂ O ₅	bdl	bdl	0.08	bdl	bdl	0.08	bdl	0.29	0.05	0.17	0.09	bdl	bdl	0.13
Cr ₂ O ₃	45.07	0.12	0.09	bdl	bdl	bdl	bdl	bdl	0.18	0.28	bdl	0.12	0.04	bdl
Fe ₂ O ₃ *	10.07	19.49	15.20	23.41	22.99	7.43	28.07	8.72	37.21	39.01	39.97	29.80	26.22	27.81
FeO*	19.22	49.59	35.30	46.85	47.37	39.11	45.04	33.55	43.95	43.04	43.10	47.22	47.24	48.85
MnO	0.06	0.63	0.38	1.11	0.96	1.21	1.48	1.45	1.69	1.58	1.61	2.20	1.77	1.69
MgO	11.07	3.06	2.78	2.87	2.81	2.37	2.39	5.06	0.30	0.26	0.04	0.29	1.40	0.02
NiO	0.18	0.04	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	0.03	bdl	bdl
ZnO	0.06	0.09	0.02	0.14	0.15	0.04	0.18	0.05	0.21	0.20	0.26	0.27	0.22	0.32
Total	101.58	99.94	99.28	100.33	99.80	99.79	99.61	97.95	100.24	99.94	100.06	100.73	100.14	99.98
<i>Formula based on 3 cations and 4 oxygens, respectively 2 cations and 3 oxygens for ilmenite</i>														
Si	0.00	0.00	0.00	0.00	0.00	bdl	0.00	bdl	0.00	0.00	0.00	0.00	0.00	0.00
Ti	0.05	0.70	0.85	0.60	0.63	0.93	0.57	0.91	0.46	0.42	0.42	0.57	0.60	0.60
Al	0.51	0.04	0.00	0.15	0.10	0.00	0.08	0.00	0.02	0.02	0.00	0.02	0.07	bdl
V	0.00	0.00	0.00	0.00	0.00	bdl	0.00	0.00	0.00	0.00	0.00	0.00	0.00	bdl
Nb	bdl	bdl	0.00	bdl	bdl	0.00	bdl	0.00	0.00	0.00	0.00	bdl	bdl	0.00
Cr	1.14	0.00	0.00	bdl	bdl	bdl	bdl	bdl	0.01	0.01	bdl	0.00	0.00	0.00
Fe ³⁺	0.24	0.54	0.29	0.64	0.63	0.14	0.78	0.16	1.06	1.11	1.14	0.84	0.73	0.79
Fe ²⁺	0.51	1.52	0.74	1.41	1.45	0.82	1.39	0.70	1.38	1.36	1.37	1.47	1.46	1.54
Mn	0.00	0.02	0.01	0.03	0.03	0.03	0.05	0.03	0.05	0.05	0.05	0.07	0.06	0.05
Mg	0.53	0.17	0.10	0.15	0.15	0.09	0.13	0.19	0.02	0.01	0.00	0.02	0.08	0.00
Ni	0.00	0.00	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	0.00	bdl	bdl
Zn	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.01	0.01	0.01	0.01	0.01
Total cations	3.00	3.00	2.00	3.00	3.00	2.00	3.00	2.00	3.00	3.00	3.00	3.00	3.00	3.00

bdl: below detection limit; *Fe₂O₃-FeO ratio was calculated.

Table 6: Selected EPMA analyses for aegirinite from the Tarosero Volcanic Complex.

Rock type Thin section	Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)		Peralkaline olivine trachyte	
	TA-054	TA-073	TA-008	TA-072	TA-005	TA-006
wt. %						
SiO ₂	39.71	42.16	39.22	40.13	41.70	41.43
TiO ₂	5.70	5.71	6.54	6.02	7.76	6.49
ZrO ₂	0.39	0.09	0.27	0.24	0.08	0.13
Al ₂ O ₃	1.60	0.45	1.09	1.46	0.19	0.28
Cr ₂ O ₃	bdl	bdl	bdl	bdl	bdl	bdl
Fe ₂ O ₃ *	11.96	9.24	11.12	10.84	5.74	8.37
FeO*	31.31	32.78	31.21	31.92	36.60	34.88
MnO	1.55	1.58	1.52	1.63	1.64	1.38
MgO	0.96	0.58	1.04	1.14	0.16	0.14
CaO	0.71	0.28	0.88	0.62	0.22	0.25
Na ₂ O	7.35	7.91	7.30	7.31	7.53	7.63
K ₂ O	0.06	0.04	0.04	0.07	0.03	0.02
F	bdl	bdl	bdl	bdl	bdl	bdl
Cl	bdl	bdl	bdl	bdl	bdl	bdl
Total	101.29	100.83	100.23	101.37	101.66	101.00
<i>Formula based on 14 cations and 20 oxygens</i>						
Si	5.61	5.94	5.60	5.65	5.88	5.88
Ti	0.61	0.61	0.70	0.64	0.82	0.69
Zr	0.03	0.01	0.02	0.02	0.01	0.01
Al	0.27	0.08	0.18	0.24	0.03	0.05
Cr	bdl	bdl	bdl	bdl	bdl	bdl
Fe ³⁺	1.27	0.98	1.20	1.15	0.61	0.89
Fe ²⁺	3.70	3.87	3.73	3.76	4.32	4.14
Mn	0.19	0.19	0.18	0.19	0.20	0.17
Mg	0.20	0.12	0.22	0.24	0.03	0.03
Ca	0.11	0.04	0.13	0.09	0.03	0.04
Na	2.01	2.16	2.02	2.00	2.06	2.10
K	0.01	0.01	0.01	0.01	0.01	0.00
Total cations	14.00	14.00	14.00	14.00	14.00	14.00
F	bdl	bdl	bdl	bdl	bdl	bdl
Cl	bdl	bdl	bdl	bdl	bdl	bdl
Total anions	bdl	bdl	bdl	bdl	bdl	bdl

bdl: below detection limit; *Fe₂O₃-FeO ratio was calculated.

Table 7: Selected EPMA analyses for amphibole from the Tarosero Volcanic Complex.

Rock type Thin section	Andesite		Latite		Trachyte		Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)		Peralkaline olivine trachyte	
	TA-044	TA-045	TA-118	TA-073	TA-078	TA-072	TA-054	TA-021	TA-008	TA-030	TA-006	TA-006
wt.%	large	interstitial	large	altered rim	large	large	interstitial	altered rim	interstitial	cumulate	interstitial	interstitial
SiO ₂	40.01	51.03	40.53	50.00	43.87	44.94	50.60	50.91	50.90	46.53	50.11	49.48
TiO ₂	6.14	1.48	5.69	1.08	3.21	2.91	0.14	1.09	1.05	2.30	1.44	3.73
ZrO ₂	0.05	0.08	0.03	bdl	0.05	bdl	0.03	bdl	0.07	0.06	0.08	0.08
Al ₂ O ₃	12.74	1.40	11.53	3.37	8.00	6.79	0.44	2.13	1.79	5.07	0.24	0.47
Cr ₂ O ₃	bdl	bdl	bdl	0.05	bdl	0.04	bdl	bdl	bdl	bdl	bdl	bdl
Fe ₂ O ₃ *	0.00	0.00	0.00	1.60	0.00	1.32	5.46	3.13	2.71	0.00	3.40	0.00
FeO*	13.51	18.85	13.81	12.99	15.09	15.52	27.28	13.07	17.50	20.62	30.45	31.29
MnO	0.30	0.61	0.36	0.78	0.56	0.71	1.59	1.00	1.02	0.92	1.76	1.35
MgO	10.16	10.98	10.82	13.85	11.85	11.74	1.95	13.21	10.63	9.98	0.51	1.34
CaO	11.81	6.46	11.27	8.06	10.12	9.93	2.59	6.54	6.01	8.58	2.26	2.31
Na ₂ O	2.89	5.38	3.15	5.25	3.90	4.13	7.63	6.07	6.18	4.05	7.89	7.89
K ₂ O	1.08	1.19	0.98	1.26	1.12	1.10	1.58	1.29	1.33	1.26	1.23	0.99
F	bdl	3.25	bdl	3.27	0.98	1.07	2.92	3.18	3.36	1.00	2.66	2.02
Cl	0.01	0.03	0.02	bdl	0.03	0.02	0.01	bdl	bdl	0.01	0.03	0.02
H ₂ O*	1.98	0.44	1.98	0.48	1.52	1.50	0.53	0.52	0.42	1.51	0.63	0.93
Total	100.68	99.80	100.15	100.66	99.87	101.25	101.51	100.79	101.55	101.45	101.57	101.04
<i>Formula based on 16 cations and 23 oxygens</i>												
Si	6.04	7.75	6.13	7.37	6.60	6.70	7.94	7.51	7.59	7.04	7.93	7.82
Ti	0.70	0.17	0.65	0.12	0.36	0.33	0.02	0.12	0.12	0.26	0.17	0.44
Zr	0.00	0.01	0.00	bdl	0.00	bdl	0.00	bdl	0.00	0.00	0.01	0.01
Al	2.27	0.25	2.05	0.58	1.42	1.19	0.08	0.37	0.31	0.90	0.05	0.09
Cr	bdl	bdl	bdl	0.01	bdl	0.00	0.00	bdl	bdl	bdl	bdl	bdl
Fe ³⁺	0.00	0.00	0.00	0.18	0.00	0.15	0.64	0.35	0.30	0.00	0.40	0.00
Fe ²⁺	1.70	2.39	1.75	1.60	1.90	1.94	3.58	1.61	2.18	2.61	4.03	4.14
Mn	0.04	0.08	0.05	0.10	0.07	0.09	0.21	0.12	0.13	0.12	0.24	0.18
Mg	2.29	2.49	2.44	3.04	2.66	2.61	0.46	2.90	2.36	2.25	0.12	0.32
Ca	1.91	1.05	1.83	1.27	1.63	1.59	0.44	1.03	0.96	1.39	0.38	0.39
Na	0.85	1.59	0.92	1.50	1.14	1.20	2.32	1.73	1.79	1.19	2.42	2.42
K	0.21	0.23	0.19	0.24	0.22	0.21	0.32	0.24	0.25	0.24	0.25	0.20
Total cations	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
F	bdl	1.55	bdl	1.53	0.47	0.51	1.45	1.49	1.58	0.48	1.33	1.01
Cl	0.00	0.01	0.00	bdl	0.01	0.00	0.00	bdl	0.00	0.00	0.01	0.00
OH	2.00	0.44	2.00	0.47	1.53	1.49	0.55	0.51	0.42	1.52	0.66	0.98
Total anions	2.00	2.00	2.00	2.00	2.00	2.00	2.00	2.00	2.00	2.00	2.00	2.00

bdl: below detection limit; *Fe₂O₃-FeO ratio and H₂O was calculated.

Table 8: Selected EPMA analyses for feldspar from the Tarosero volcanic complex.

Rock type	Basalt		Andesite		Latite		Trachyte		Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)		Peralkaline olivine trachy				
Thin section	TA-042	TA-042	TA-044	TA-044	TA-046	TA-046	TA-118	TA-116	TA-078	TA-054	TA-035	TA-021	TA-022	TA-030	TA-021	TA-006	TA-005
Mineral	afs	pl	pl	pl	pl	afs	pl	afs	afs	afs	afs	afs	afs	afs	pl	afs	afs
wt. %	small	small	small	small	small	small	large	small	large	large	small	large	small	small	xenolith	large	small
SiO ₂	64.58	56.60	59.18	57.97	58.64	64.12	60.90	65.98	64.23	67.31	66.72	67.01	67.40	65.73	62.91	66.89	67.22
TiO ₂	0.34	0.19	0.28	0.08	0.09	0.14	0.06	0.13	0.07	0.04	0.05	0.08	0.02	0.06	0.08	0.05	0.06
Al ₂ O ₃	19.86	26.32	24.07	25.54	25.28	20.95	24.21	19.57	21.45	19.07	18.74	19.04	18.86	20.83	22.78	19.07	19.07
Fe ₂ O ₃	0.71	1.15	1.09	0.39	0.54	0.53	0.31	0.51	0.25	0.39	0.46	0.51	0.50	0.16	0.29	0.25	0.48
BaO	0.06	0.06	0.15	0.08	0.11	0.33	0.24	bdl	0.30	0.05	0.07	0.03	bdl	0.52	0.16	0.15	0.09
SrO	bdl	0.07	0.59	0.30	0.31	0.09	0.72	bdl	0.54	bdl	bdl	bdl	bdl	0.08	0.13	bdl	bdl
CaO	1.44	10.26	8.69	7.87	7.33	2.49	5.41	0.70	2.39	0.21	0.23	0.22	0.04	1.67	4.02	0.27	0.15
Na ₂ O	6.61	5.30	5.98	6.78	6.63	7.95	7.66	6.99	8.68	8.19	7.77	7.71	7.58	8.32	8.30	7.49	8.06
K ₂ O	6.31	0.80	0.53	0.61	0.89	3.09	1.05	6.06	2.26	5.01	5.77	5.65	6.12	3.18	1.28	5.82	4.89
Total	99.91	100.74	99.93	99.62	99.83	99.68	100.55	99.95	100.18	100.27	99.81	100.25	100.53	100.54	99.94	99.99	100.02

Formula based on 8 oxygens	
Si	2.91
Ti	0.01
Al	1.05
Fe ³⁺	0.02
Ba	0.00
Sr	bdl
Ca	0.07
Na	0.58
K	0.36
Total cations	5.01

afs: alkali feldspar; pl: plagioclase; bdl: below detection limit

Table 9: Selected EPMA analyses for sodalite group minerals from the Tarosero Volcanic Complex.

Rock type	Peralkaline trachyte/phonolite		Peralkaline phonolite (agpaitic)	
Thin section	TA-054	TA-035	TA-022	TA-008_2
<i>wt. %</i>				
SiO ₂	41.37	38.62	42.19	42.44
Al ₂ O ₃	23.12	28.49	21.78	21.86
Fe ₂ O ₃	5.60	1.87	6.00	5.48
CaO	bdl	bdl	bdl	bdl
Na ₂ O	23.71	25.35	24.62	21.90
K ₂ O	0.34	0.15	0.12	0.76
SO ₃	bdl	bdl	0.06	bdl
Cl	6.85	6.99	6.77	6.86
Total	99.45	99.89	100.01	97.75
<i>Formula based on 25 oxygens</i>				
Si	6.75	6.25	6.86	7.01
Al	4.45	5.44	4.17	4.25
Fe ³⁺	0.69	0.23	0.73	0.68
Ca	bdl	bdl	bdl	bdl
Na	7.50	7.96	7.76	7.01
K	0.07	0.03	0.02	0.16
Total cations	19.47	19.91	19.56	19.11
SO ₃	bdl	bdl	0.01	bdl
Cl	1.90	1.92	1.87	1.92
Total anions	1.90	1.92	1.87	1.92

bdl: below detection limit

Table 10: Selected EPMA analyses for eudialyte group minerals from the Tarosero volcanic complex.

Rock type Thin section	Peralkaline phonolite (agpaitic)					
	TA-030	TA-072	TA-030	TA-022	TA-030	Dawson, 1997
<i>wt. %</i>						
SiO ₂	52.51	52.03	53.16	52.03	51.43	50.10
TiO ₂	0.29	0.22	0.37	0.22	0.41	0.18
ZrO ₂	11.01	11.27	9.42	9.75	8.80	11.30
HfO ₂	0.30	0.26	0.13	0.26	0.13	bdl
Al ₂ O ₃	0.09	0.17	0.51	0.18	0.16	0.20
FeO	7.01	7.60	6.63	7.75	6.77	7.50
MnO	1.23	1.56	1.25	1.52	1.25	1.24
CaO	8.34	7.83	7.23	8.25	8.25	8.70
Na ₂ O	13.59	13.88	11.58	12.23	13.44	12.90
K ₂ O	0.33	0.31	0.71	0.33	0.34	0.33
Nb ₂ O ₅	0.54	0.52	0.90	0.71	1.01	0.55
SrO	0.17	bdl	bdl	0.18	bdl	bdl
La ₂ O ₃	0.59	0.48	0.84	0.59	0.62	0.57
Ce ₂ O ₃	1.13	1.15	1.74	1.17	1.40	0.94
Nd ₂ O ₃	0.38	0.37	0.65	0.44	0.49	0.58
Y ₂ O ₃	0.38	0.37	0.66	0.50	0.69	bdl
Cl	2.19	2.05	2.07	2.19	2.27	2.13
Total	99.57	99.60	97.37	97.78	96.95	96.74
<i>Formula based on Si+Al+Zr+Ti+Hf+Nb=29 cations</i>						
Si	26.02	25.93	26.10	26.22	26.30	25.84
Ti	0.11	0.08	0.14	0.08	0.16	0.07
Zr	2.66	2.74	2.26	2.40	2.19	2.84
Hf	0.04	0.04	0.02	0.04	0.02	bdl
Al	0.05	0.10	0.29	0.11	0.09	0.12
Fe ²⁺	2.90	3.17	2.72	3.27	2.90	3.23
Mn	0.52	0.66	0.52	0.65	0.54	0.54
Ca	4.43	4.18	3.80	4.45	4.52	4.81
Na	13.06	13.41	11.02	11.95	13.33	12.90
K	0.21	0.19	0.44	0.21	0.22	0.22
Nb	0.12	0.12	0.20	0.16	0.23	0.13
Sr	0.05	bdl	bdl	0.05	bdl	bdl
La	0.11	0.09	0.15	0.11	0.12	0.11
Ce	0.20	0.21	0.31	0.22	0.26	0.18
Nd	0.07	0.07	0.11	0.08	0.09	0.11
Y	0.10	0.10	0.17	0.13	0.19	bdl
Total cations	50.64	51.07	48.26	50.12	51.16	51.09
Cl	1.84	1.73	1.72	1.87	1.97	1.86

bdl: below detection limit

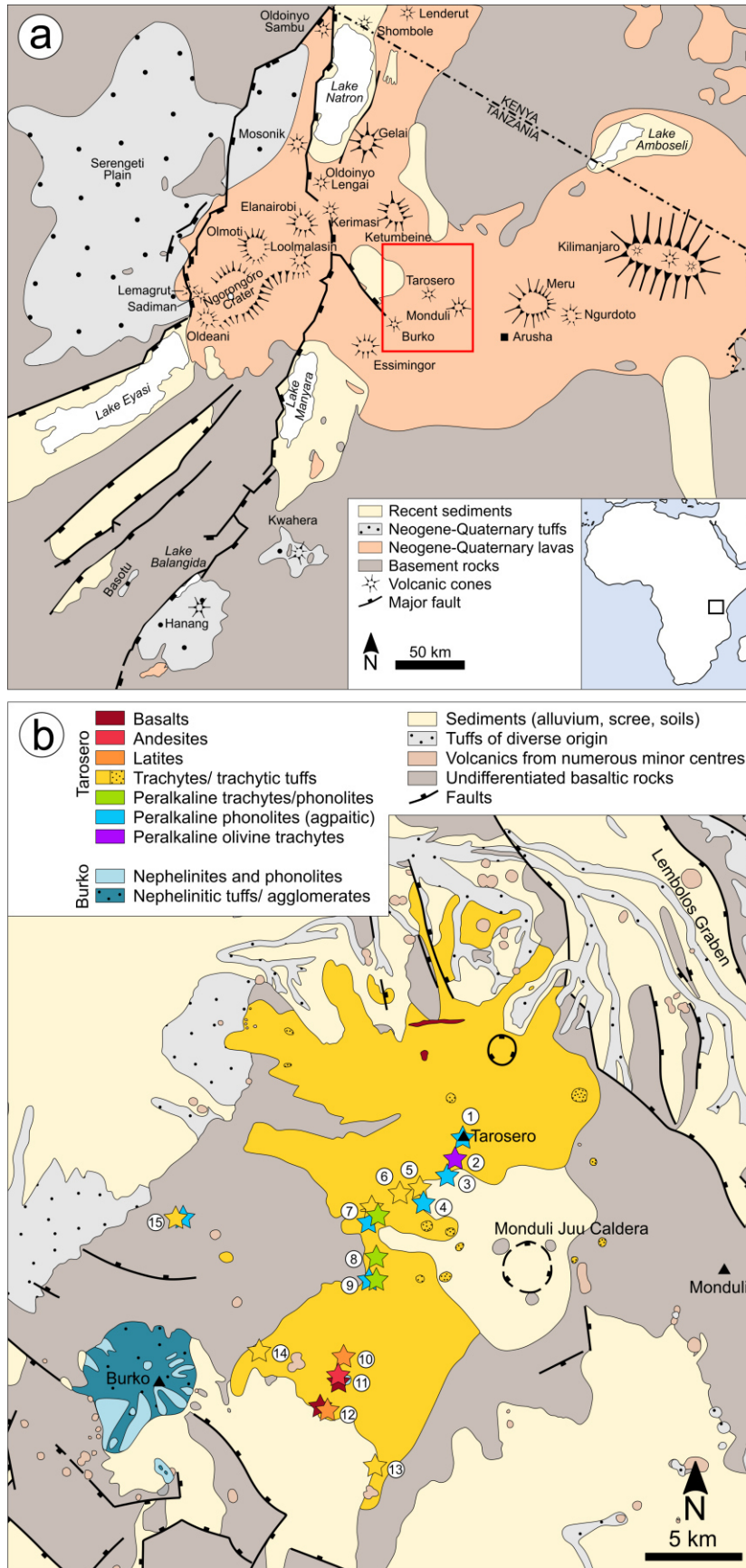


Fig. 1: Geology of northern Tanzania. a) North Tanzanian Divergence Zone (modified after Woolley, 2001). Red square represents the study area. b) Geological map of the Tarosero Volcanic Complex (modified after Dawson & Pickering, 1964), including the sample localities (indicated by the stars). The numbers refer to the samples listed in Table 1.

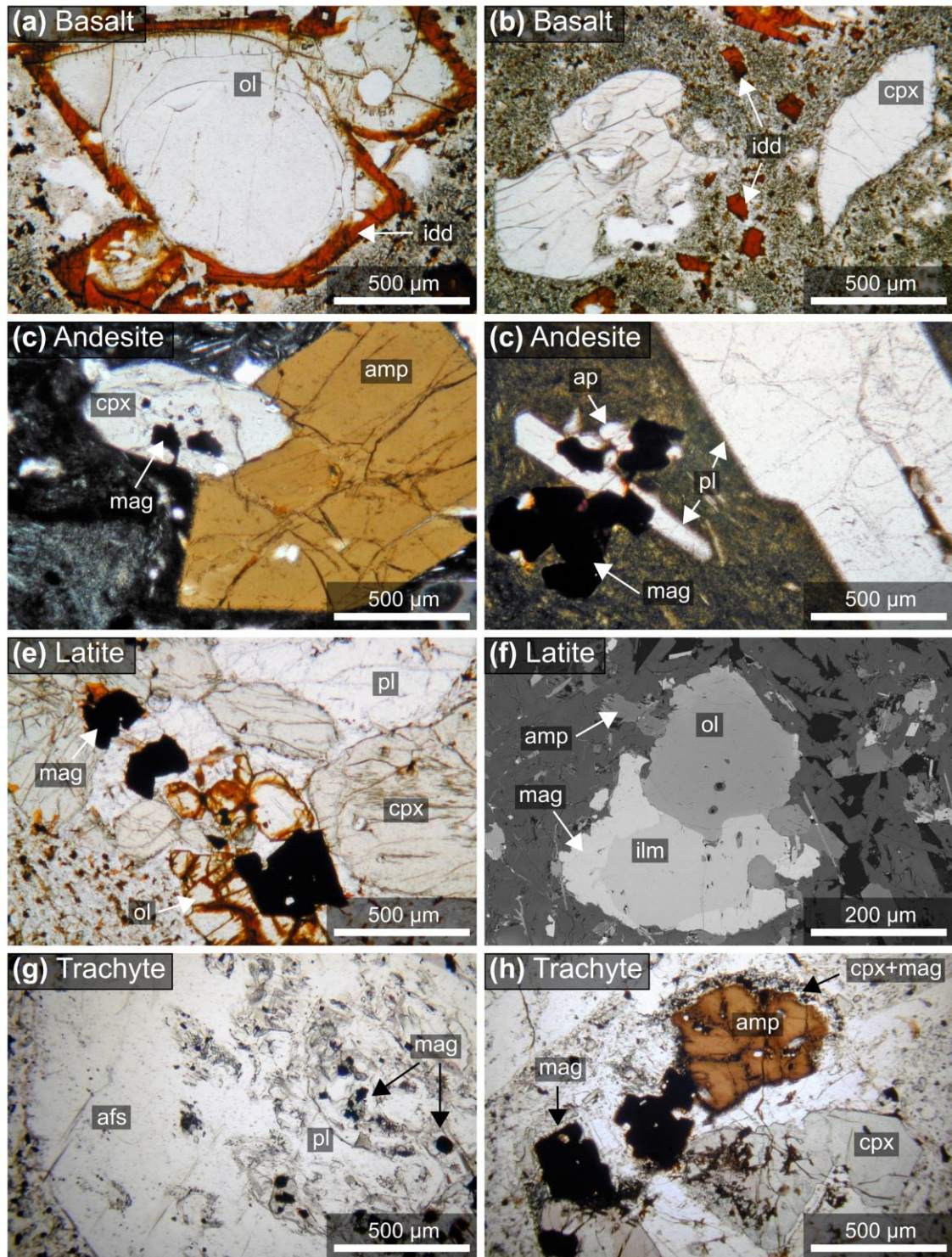


Fig. 2: Mineral textures in the primitive to intermediate rock types from Tarosero. a) Basalt with euhedral olivine (ol) crystals that are surrounded by iddingsite (idd). b) Basalt with rounded clinopyroxene (cpx) macrocrysts and small altered (to iddingsite) olivine grains. c) Andesite consisting of large euhedral amphibole (amp), and clinopyroxene with inclusions of magnetite (mag). d) Andesite showing magnetite and plagioclase (pl) phenocrysts, and small apatite (ap) in a very fine-grained groundmass. e) Latite with olivine, clinopyroxene, magnetite and plagioclase. f) Latite containing olivine, ilmenite rimmed by magnetite, and small anhedra amphibole grains. g) Trachyte showing a sieved-textured plagioclase surrounded by alkali feldspar (afs). h) Trachyte with subhedral clinopyroxene, magnetite and a dark brown amphibole that is partly rimmed by an alteration zone of fine-grained clinopyroxene and magnetite.

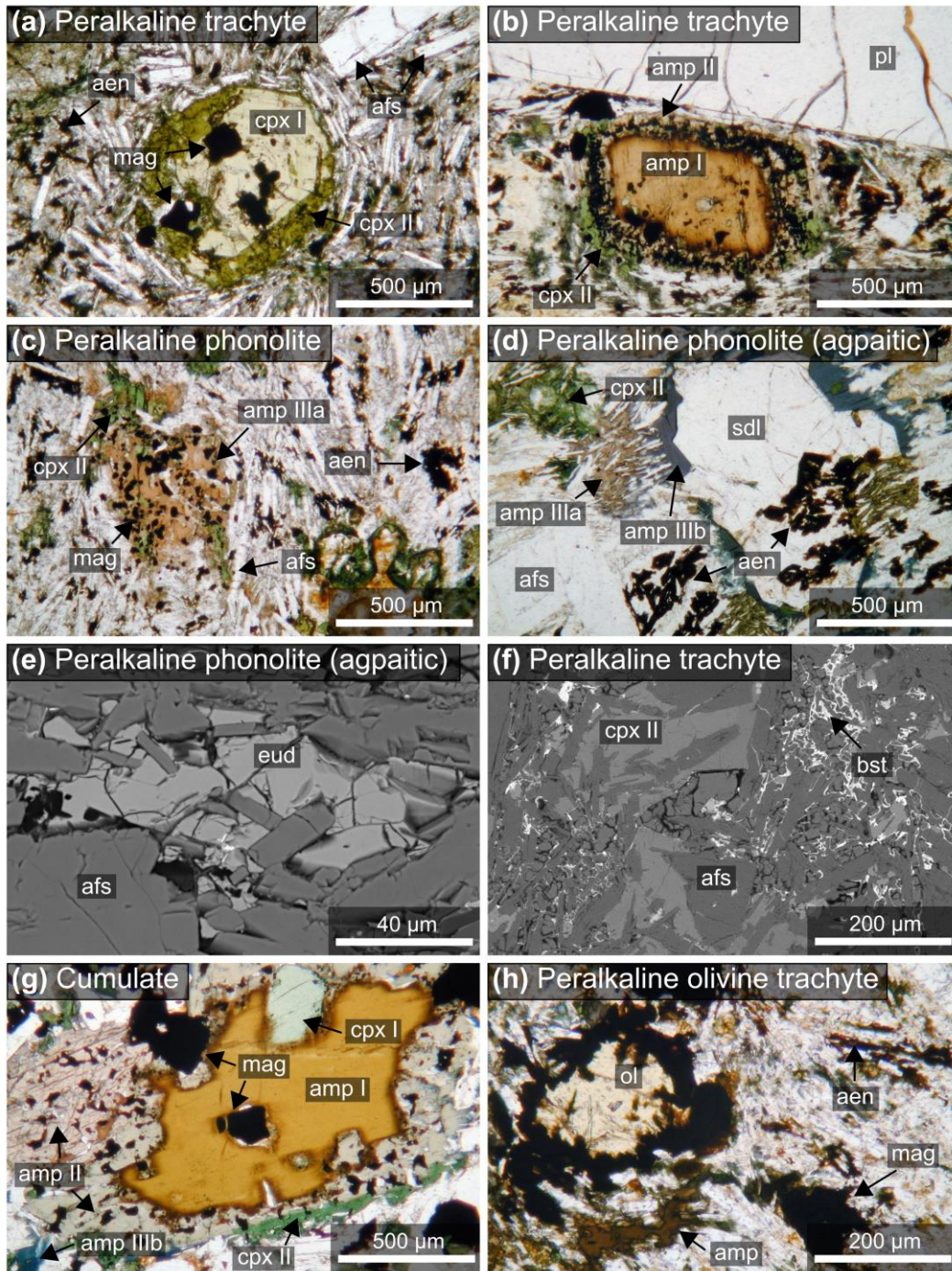


Fig. 3: Mineral textures in evolved rock types from Tarosero. a) Peralkaline trachyte containing bright green clinopyroxene (cpx I) with magnetite (mag) inclusions, surrounded by a dark green clinopyroxene generation (cpx II). The groundmass mainly consists of alkali feldspar laths and small aenigmatite (aen) grains. b) Peralkaline trachyte with a plagioclase (pl) crystal, and dark amphibole (amp I) which is rimmed by bright brown amphibole (amp II), dark green clinopyroxene and magnetite. c) Peralkaline phonolite with bright brown interstitial amphibole IIIa, dark green clinopyroxene II, aenigmatite and alkali feldspar laths. d) Peralkaline phonolite showing alkali feldspar laths, and a subhedral sodalite (sdl) grain with inclusions of aenigmatite, which is surrounded by blue amphibole IIIb. e) Peralkaline phonolite with interstitial eudialyte (eud). f) Peralkaline trachyte with interstitial bastnäsite (bst). g) Cumulate in a peralkaline phonolite consisting of dark brown anhedral amphibole I, surrounded by bright brown amphibole II, dark green clinopyroxene and blue amphibole IIIb. h) Peralkaline olivine trachyte with partly altered olivine and a fine-grained groundmass of magnetite, amphibole and aenigmatite.

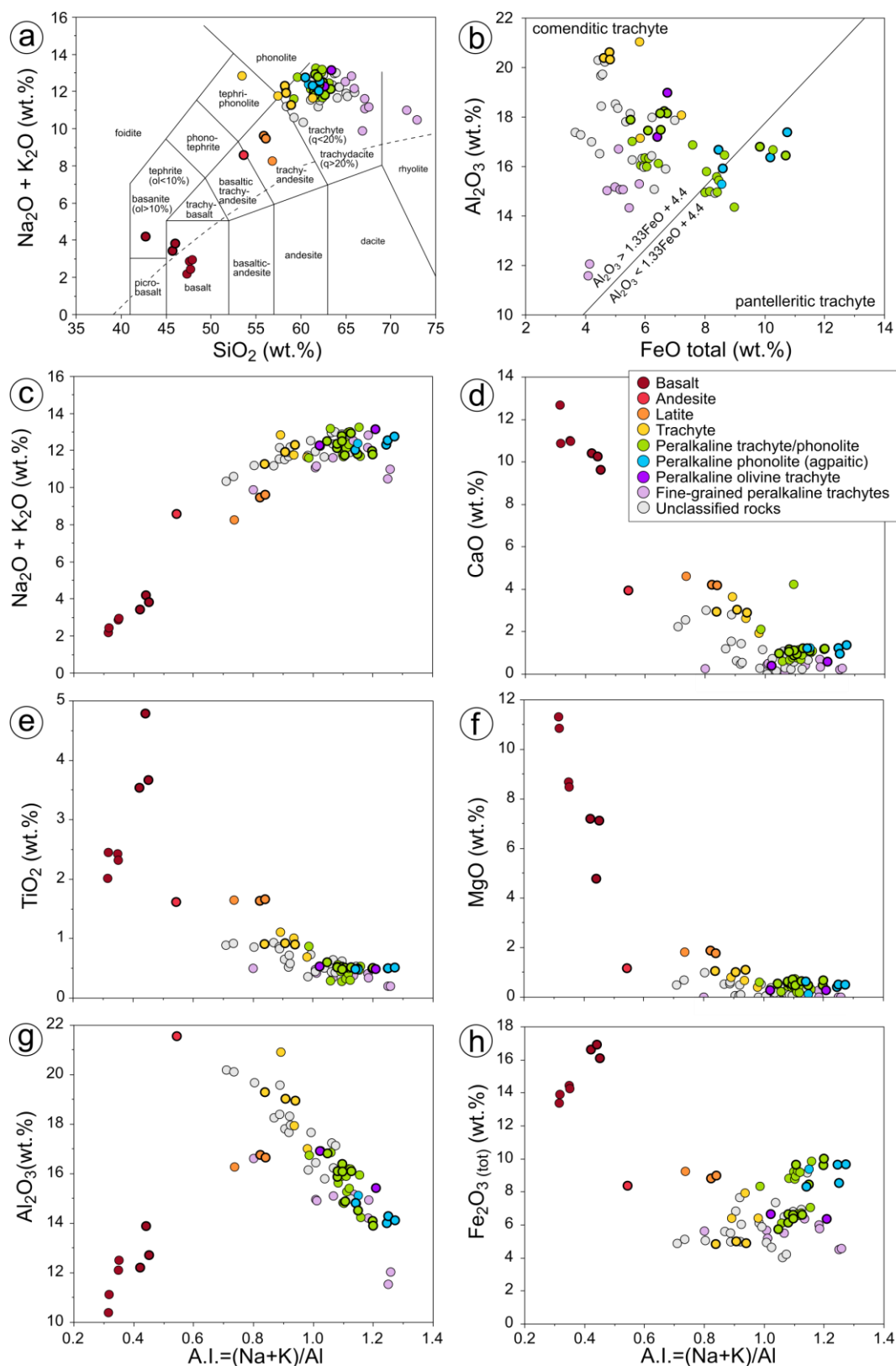


Fig. 4: Whole-rock major element composition for the rocks from Tarosero. a) TAS diagram showing the total alkalis ($\text{Na}_2\text{O}+\text{K}_2\text{O}$) vs SiO_2 (in wt. %). The dashed line divides the alkaline field (high alkali content) from the subalkaline field (according to Irvine & Baragar, 1971). b) Concentration (in wt. %) of Al_2O_3 vs total FeO (all Fe expressed as FeO). c-h) Concentration (in wt. %) of $\text{Na}_2\text{O}+\text{K}_2\text{O}$ (c), CaO (d), TiO_2 (e), MgO (f), Al_2O_3 (g) and Fe_2O_3 (h) vs the alkalinity index A.I. (molar $[\text{Na}+\text{K}]/\text{Al}$). Bold-rimmed data are from this study, literature data are from Cooper (1972) and Mana *et al.* (2015).

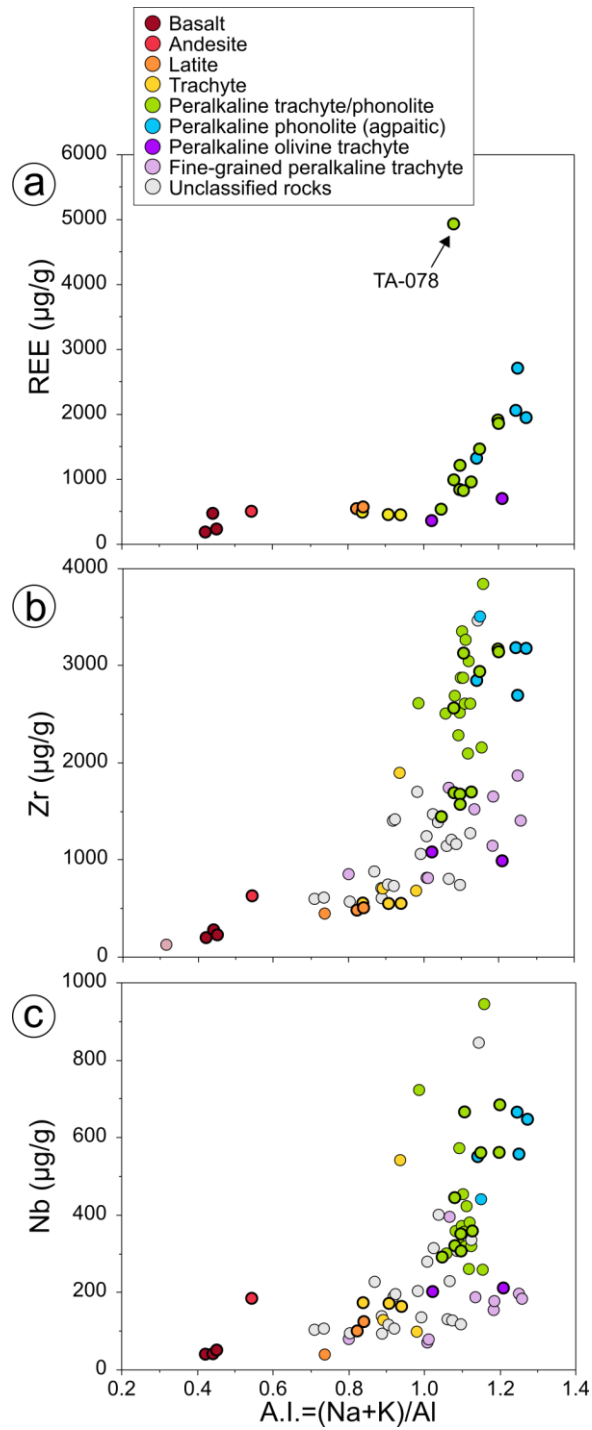


Fig. 5: Whole-rock trace element concentrations for the rocks from Tarosero. a) Concentration (in $\mu\text{g/g}$) of REE (a), Zr (b) and Nb (c) vs the alkalinity index A.I. (molar $[\text{Na}+\text{K}]/\text{Al}$). Sample TA-078 contains large amounts of REE-F-carbonates (bastnäsite). Bold-rimmed data are from this study, literature data are from Cooper (1972) and Mana *et al.* (2015).

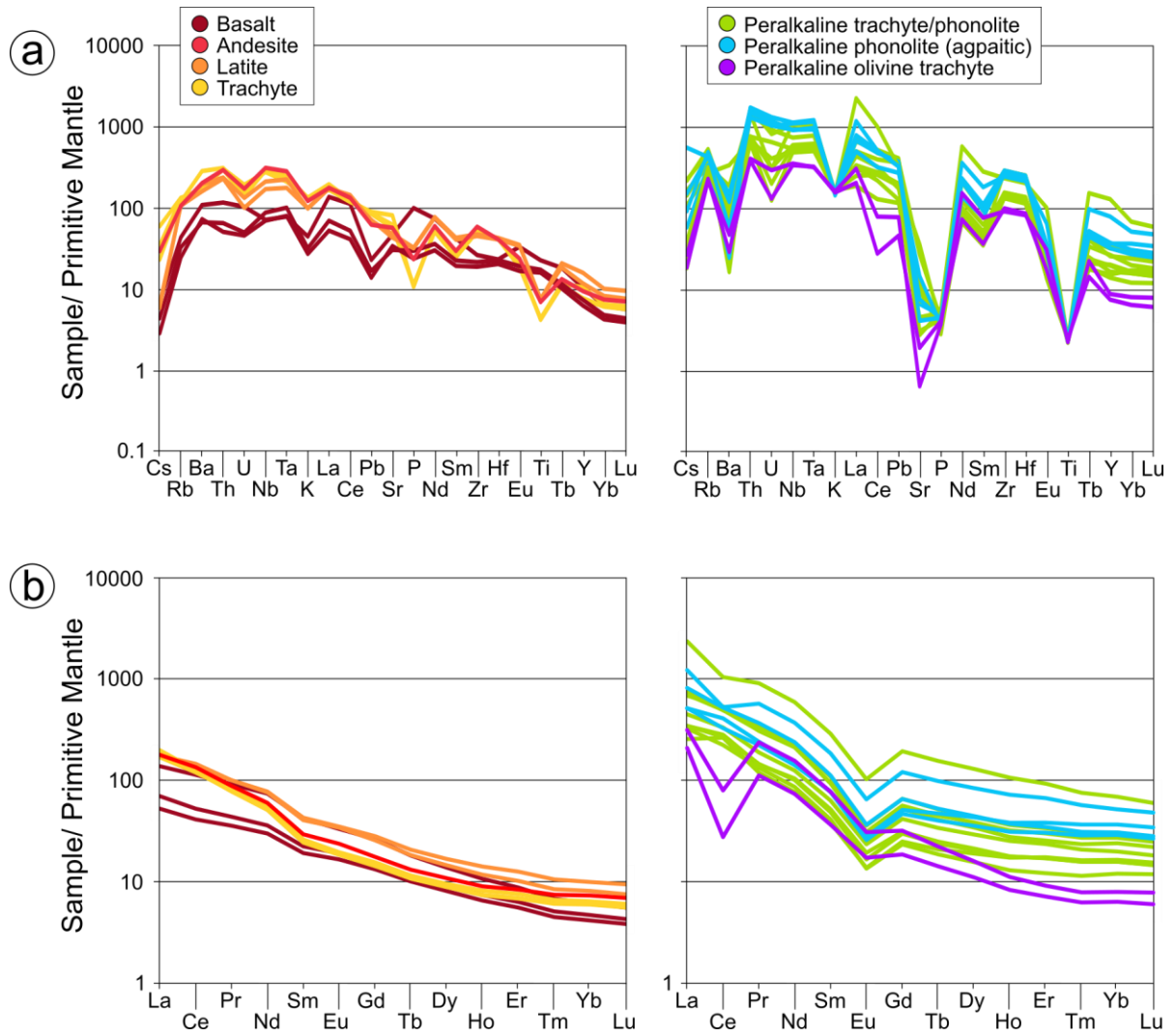


Fig. 6: Primitive mantle-normalized element pattern for the rocks from Tarosero. a) Incompatible element variation diagram. b) Rare earth elements pattern diagram. Normalization values are from Palme & O'Neill (2003).

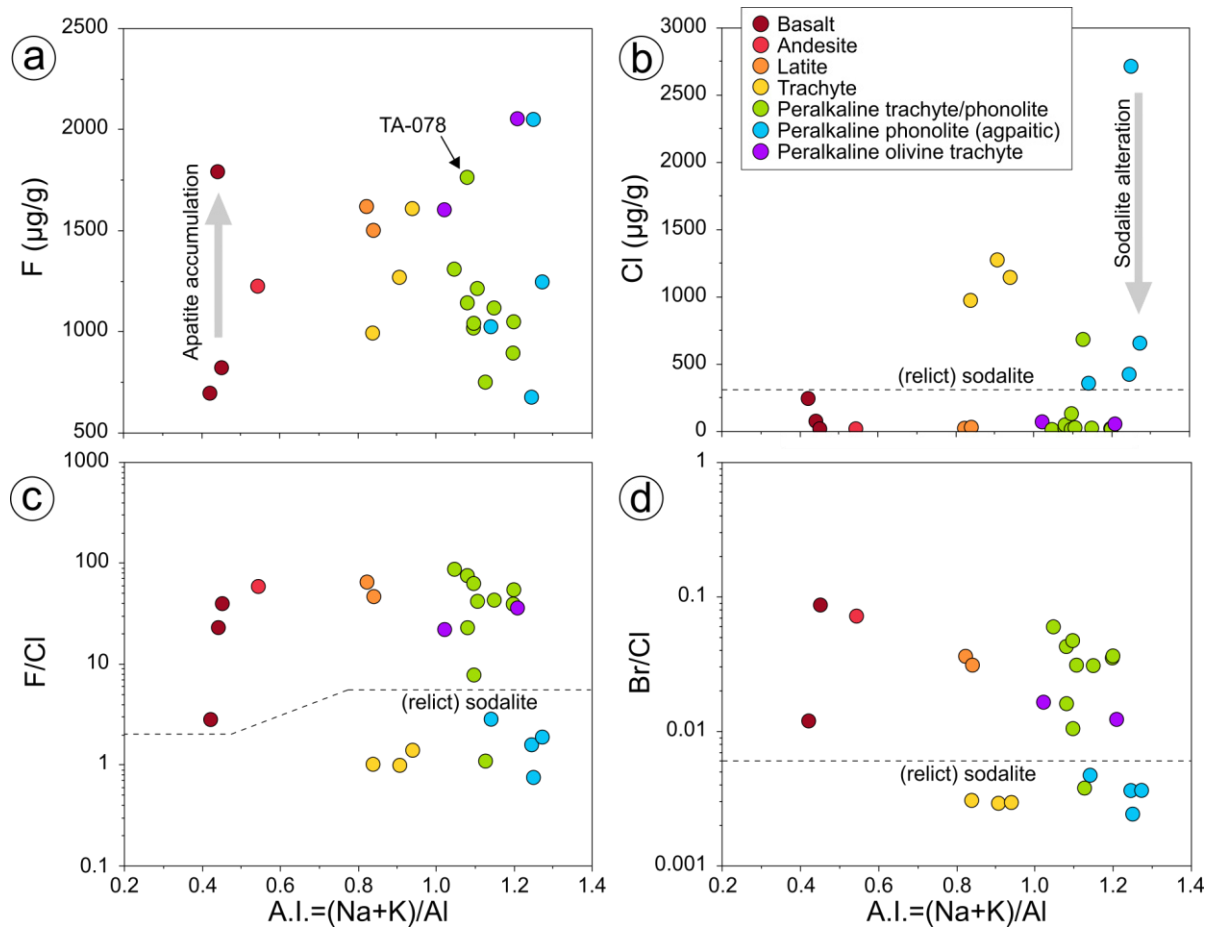


Fig. 7: Halogen composition for the rocks from Tarosero. a-b) Concentration (in $\mu\text{g/g}$) of fluorine (a) and chlorine (b) vs the alkalinity index A.I.. c-d) Ratio of F/Cl and Br/Cl vs the alkalinity index A.I.. Apatite accumulation explains the large variation in the F concentration of the basalts. The presence of relict sodalite conforms with high Cl concentrations, respectively low F/Cl and Br/Cl ratios.

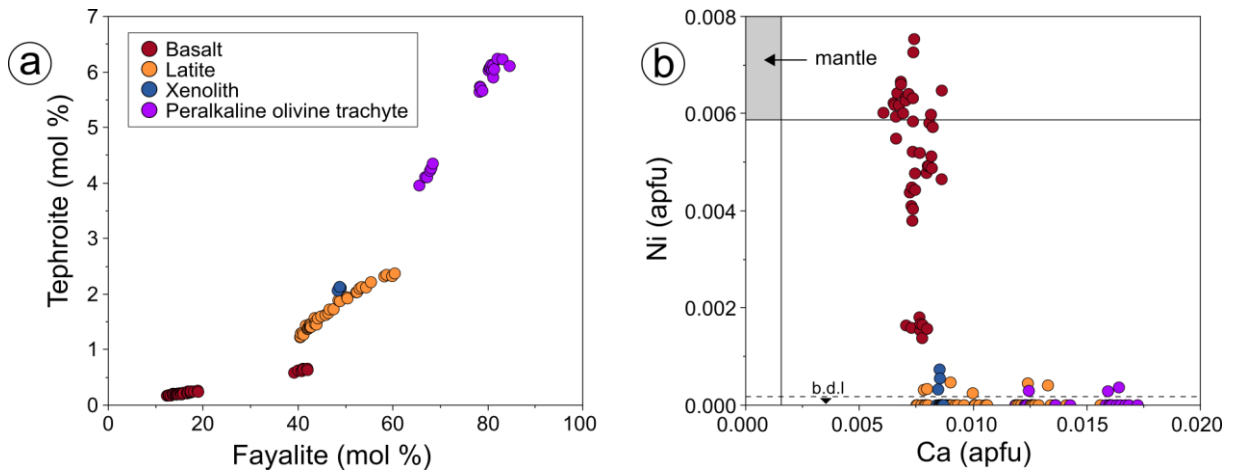


Fig. 8: Olivine composition from the rocks from Tarosero. a) The tephroite component (in mol %) positively correlates with fayalite (mol %). b) Concentration (in apfu) of Ni vs Ca. The grey field indicates the typical mantle olivine composition (Foley *et al.*, 2013, Bussweiler *et al.*, 2015).

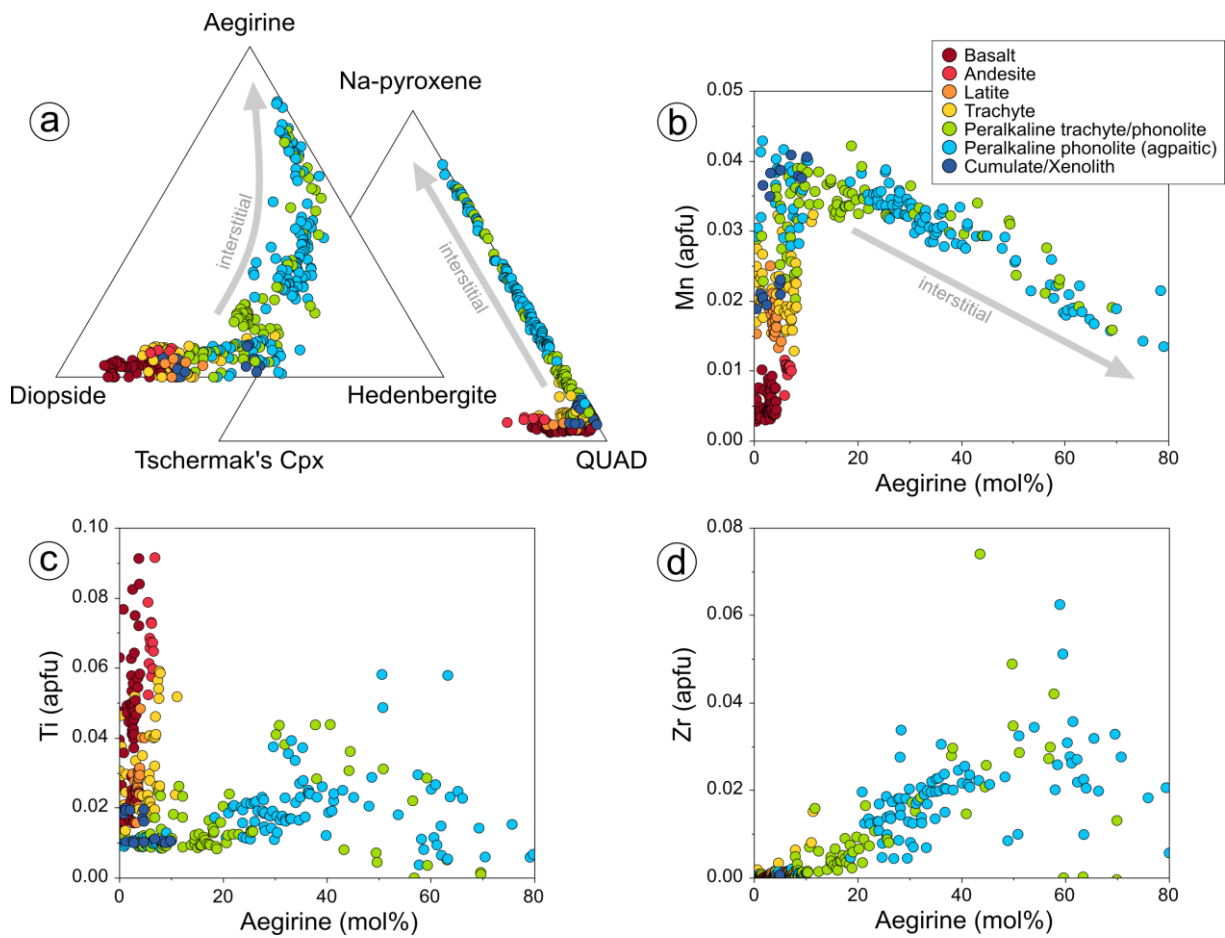


Fig. 9: Clinopyroxene composition from the rocks from Tarosero. a) Diopside ($\text{CaMgSi}_2\text{O}_6$)-aegirine ($\text{NaFeSi}_2\text{O}_6$)-hedenbergite ($\text{CaFeSi}_2\text{O}_6$) and Tschermak's clinopyroxene (CaXXSiO_6)-Na-pyroxene ($\text{NaFeSi}_2\text{O}_6 + \text{NaAlSi}_2\text{O}_6$ -QUAD ($\text{CaMgSi}_2\text{O}_6 + \text{CaFeSi}_2\text{O}_6$) triangles. b-d) Concentration (in apfu) of Mn (b), Ti (c) and Zr (d) vs the aegirine component (mol %). Grey arrows highlight the evolution trend of the interstitial clinopyroxenes.

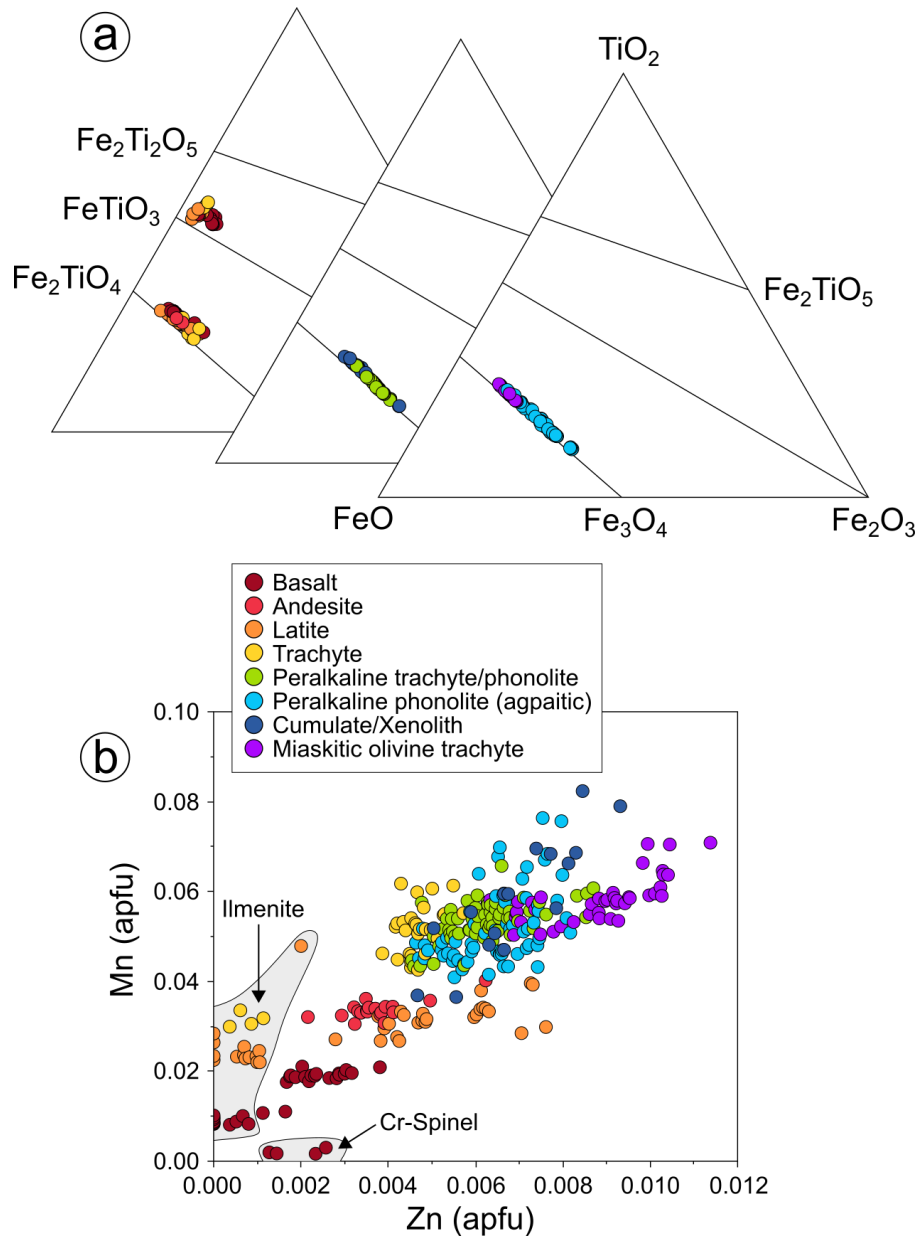


Fig. 10: Oxide composition from the rocks from Tarosero. a) Ternary FeO-TiO₂-Fe₂O₃ (molar) diagram illustrating the composition of (Ti)-magnetite and ilmenite. b) Correlation diagram of Mn (apfu) vs Zn (apfu). Grey fields indicate the compositional range for ilmenite and Cr-spinel.

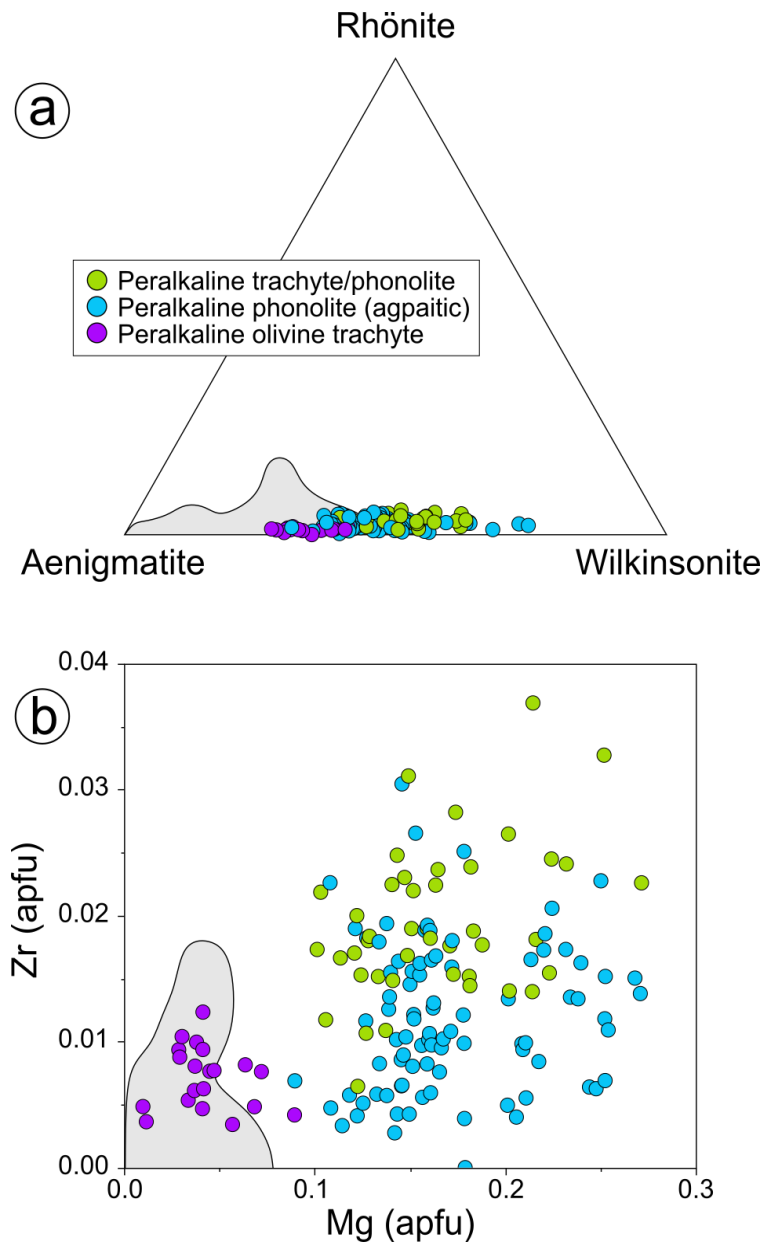


Fig. 11: Aenigmatite group composition from the rocks from Tarosero. a) Aenigmatite ($\text{Na}_2\text{Fe}_5\text{TiSi}_6\text{O}_{20}$)-Rhönite ($\text{Ca}_2[\text{Mg,Fe,Ti}]_6[\text{Si,Al}]_6\text{O}_{20}$)-Wilkinsonite ($\text{Na}_2\text{Fe}_6\text{Si}_6\text{O}_{20}$) triangle showing the compositional range of aenigmatite group minerals. b) Correlation diagram of Zr (apfu) vs Mg (apfu). Grey fields represent aenigmatite data from Ilímaussaq and Puklen (Gardar Province, Greenland; Braunger, 2016).

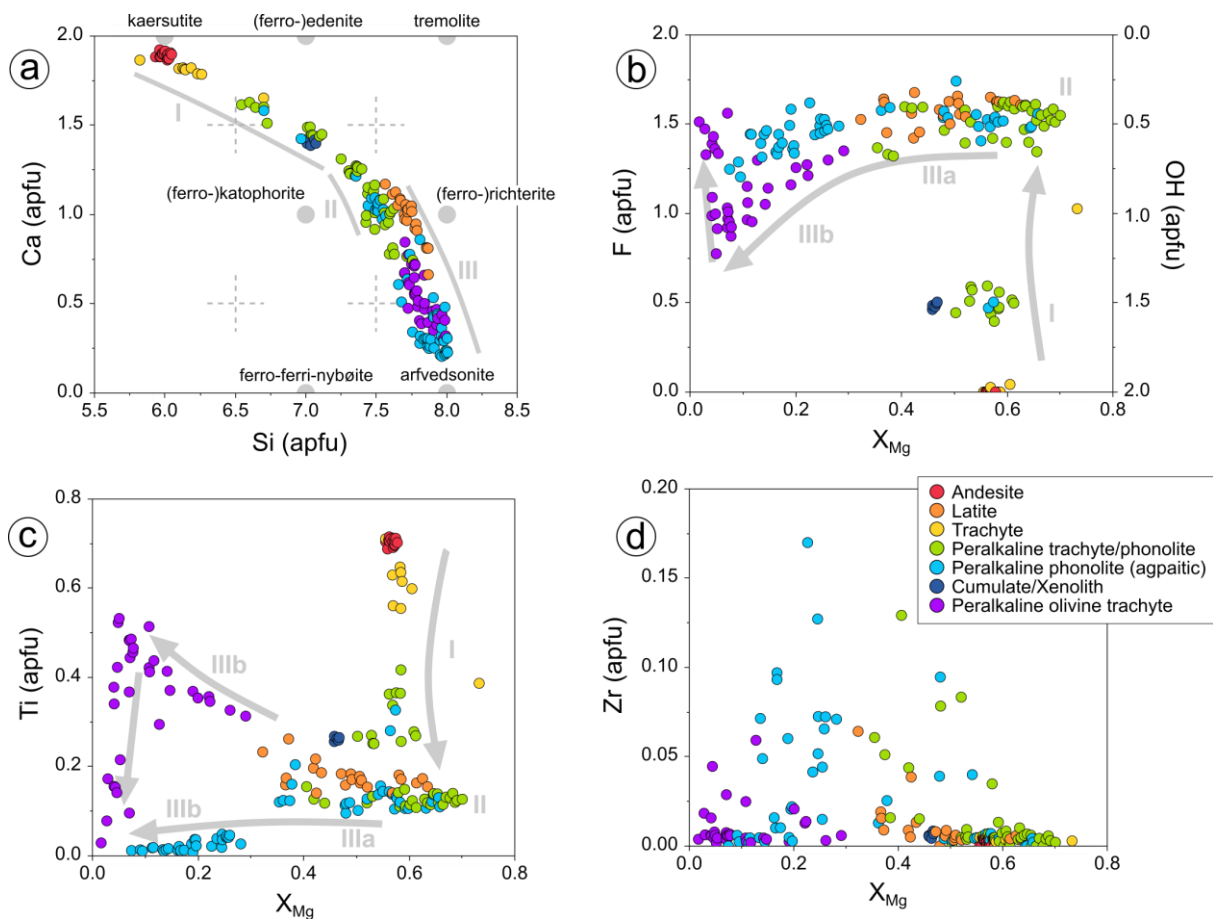


Fig. 12: Amphibole composition from the rocks from Tarosero. a) Classification (according to Hawthorne *et al.*, 2012) and correlation diagram illustrating the Ca vs Si concentration (in apfu). b-d) Concentration of Ti (b), Zr (c), F and OH (d) vs X_{Mg} ($=Mg/[Mg+Fe^{2+}]$). The grey arrows indicate evolution paths for the amphibole generations I to III.

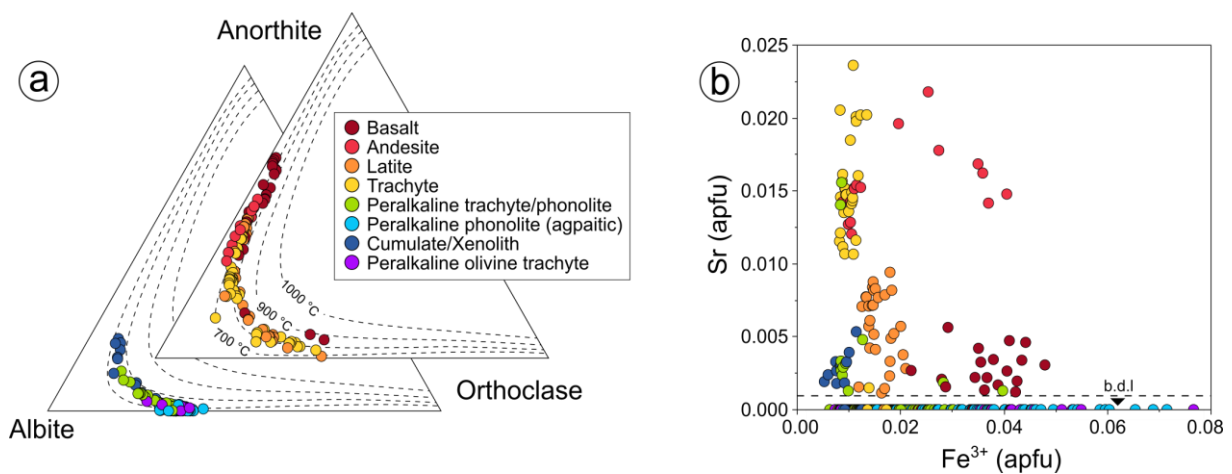


Fig. 13: Feldspar composition from the rocks from Tarosero. a) Ternary albite-anorthite-orthoclase diagram. Isotherms (dashed lines) according to Fuhrman & Lindsley (1988) at a pressure of 1 bar. b) Correlation diagram illustrating the concentration of Sr (apfu) vs Fe^{3+} (apfu).

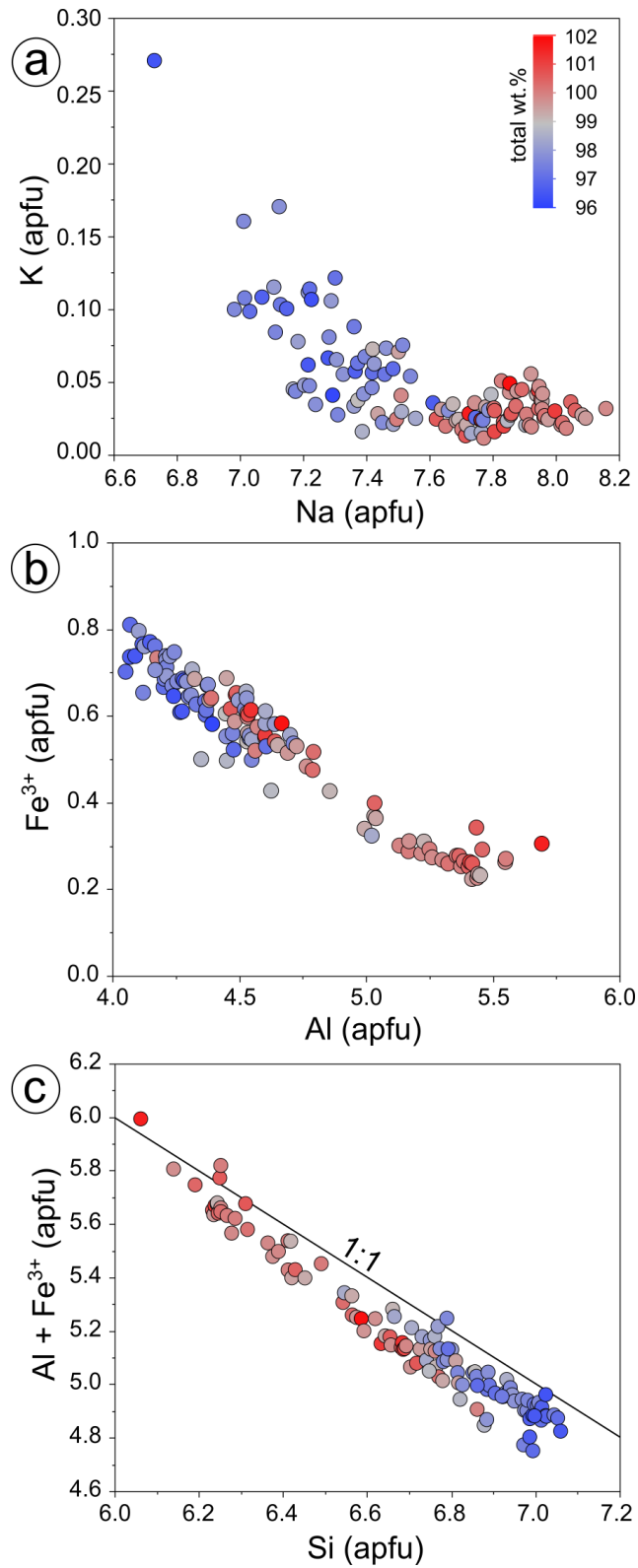


Fig. 14: Sodalite composition of the peralkaline trachytes and peralkaline phonolites from Tarosero. a-c) Correlation diagrams illustrating the K vs Na concentration (a), the Fe^{3+} vs Al concentration (b) and the $\text{Al} + \text{Fe}^{3+}$ vs Si concentration (c). The variable weight percent totals of the analyses are highlighted by the colour gradient.

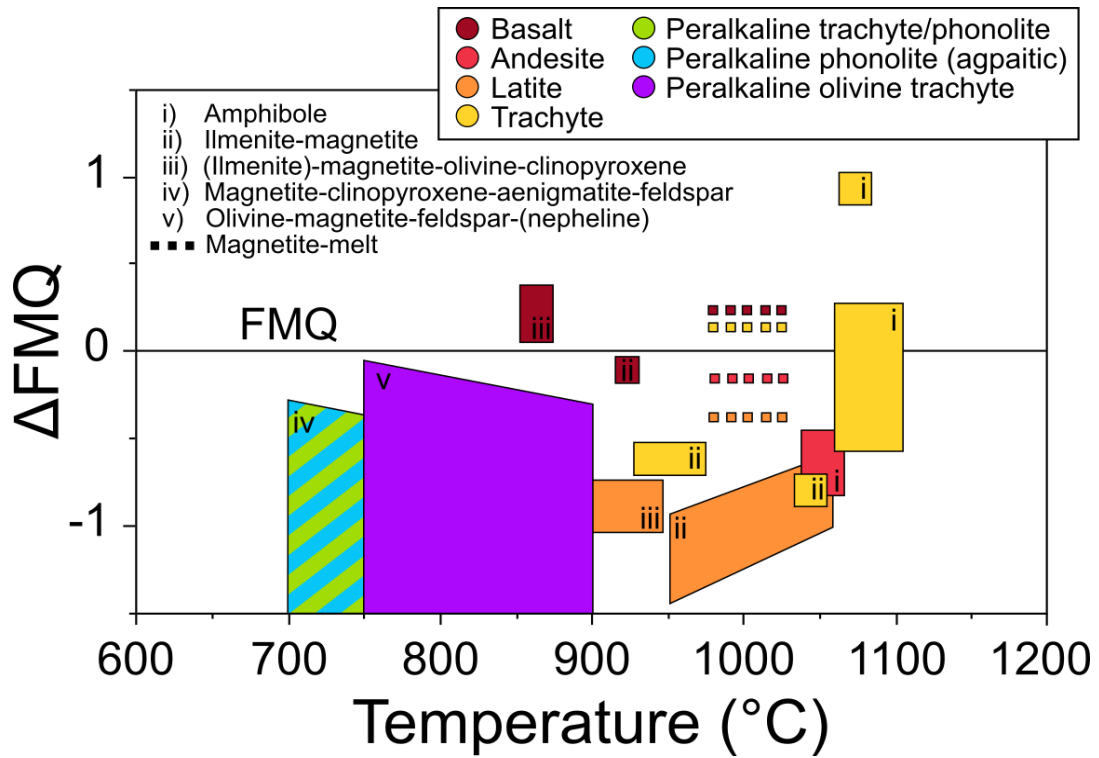


Fig. 15: Estimated temperature vs redox conditions (expressed as ΔFMQ) for the rocks from Tarosero. The numbers i-v) indicate the applied thermo- and oxybarometers. The dotted lines refer to the magnetite-melt oxybarometer of Arató & Audétat (2017), whereas the corresponding temperature has been chosen arbitrarily. See text for further information.

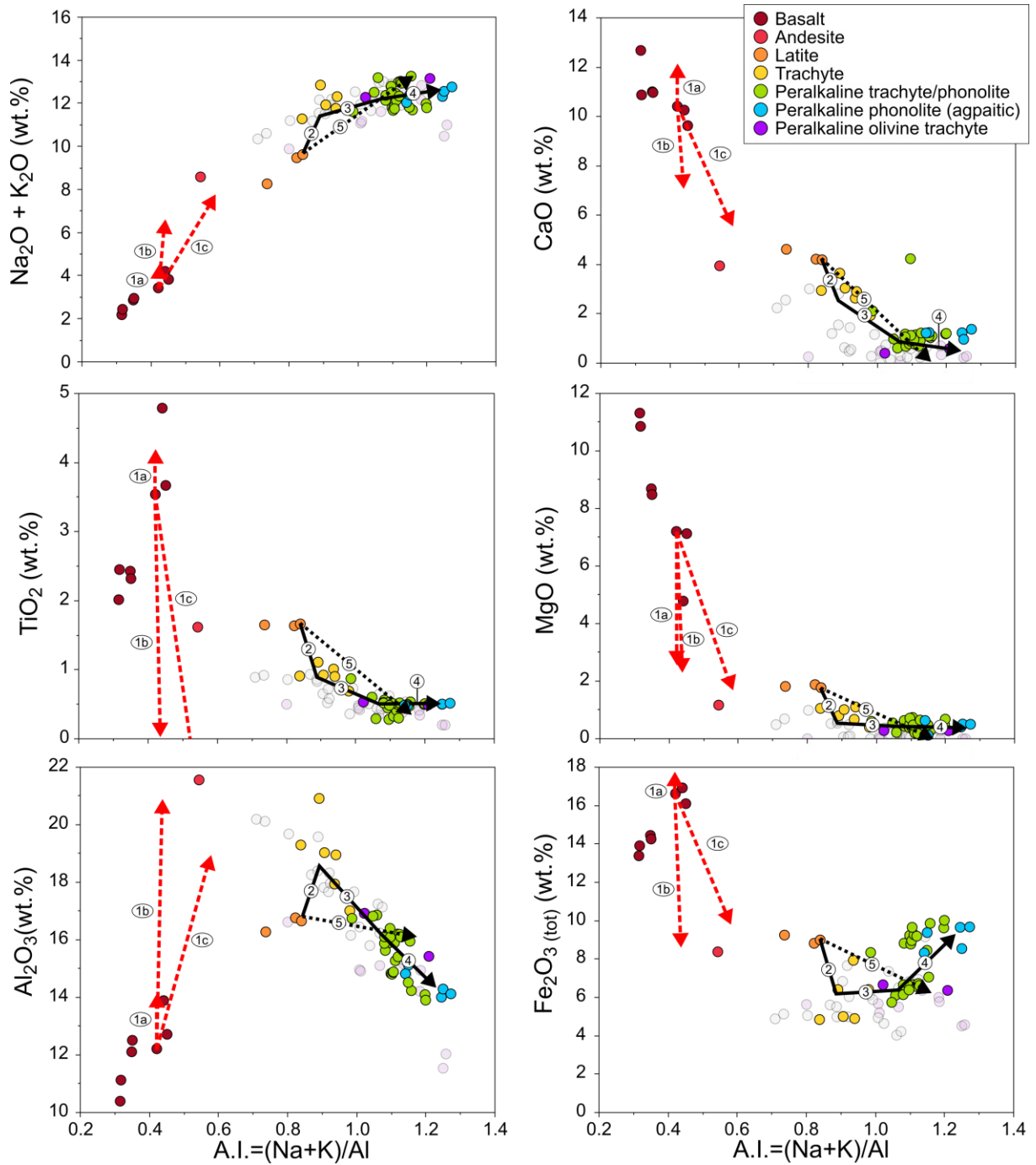


Fig. 16: Fractionation path for the rocks from Tarosero. Numbers refer to Supplementary Table 3. See text for more information.

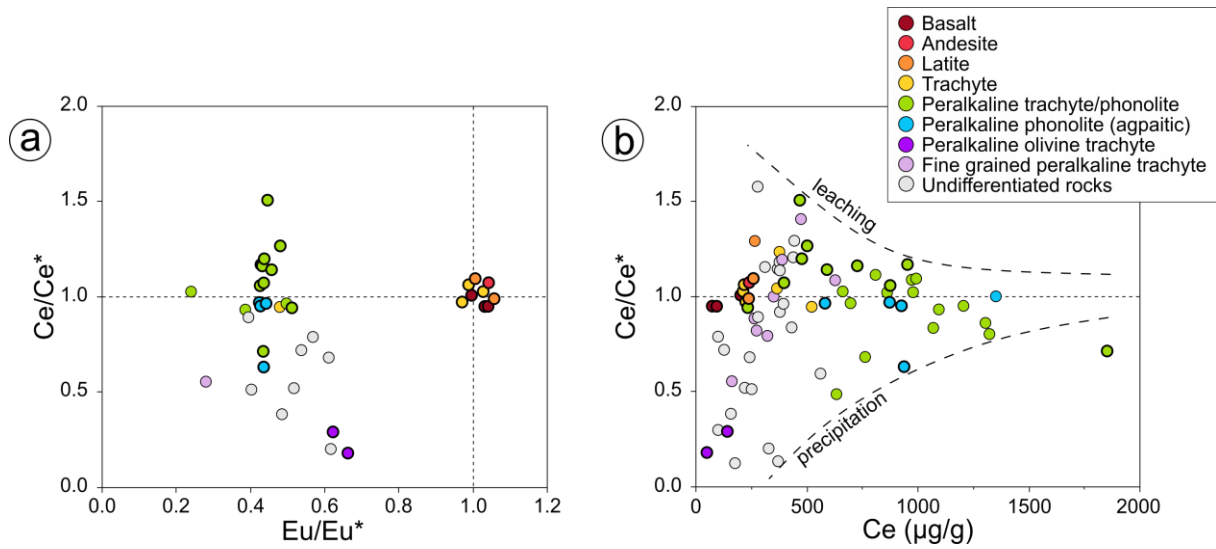


Fig. 17: Ce anomaly vs Eu anomaly (a) and Ce anomaly vs Ce concentration (in $\mu\text{g/g}$) for the rocks from Tarosero. Anomalies were calculated as $\text{Ce/Ce}^* = \text{Ce}_N / (\text{La}_N * \text{Pr}_N)^{0.5}$, respectively $\text{Eu/Eu}^* = \text{Eu}_N / (\text{Sm}_N * \text{Gd}_N)^{0.5}$.

Appendix III

Accepted publication

Study 3

Braunger, S., Marks, M.A.W., Walter, B.F., Neubauer, R., Reich, R., Wenzel, T., Parsapoor, A., Markl, G., 2018. The Petrology of the Kaiserstuhl Volcanic Complex, SW Germany: The Importance of Metasomatized and Oxidized Lithospheric Mantle for Carbonatite Generation. *Journal of Petrology* 59, 1731-1762. <http://dx.doi.org/10.1093/petrology/egy078>.

The Petrology of the Kaiserstuhl Volcanic Complex, SW Germany: The Importance of Metasomatized and Oxidized Lithospheric Mantle for Carbonatite Generation

S. Braunger *, M. A. W. Marks, B. F. Walter, R. Neubauer, R. Reich, T. Wenzel, A. Parsapoor and G. Markl

Fachbereich Geowissenschaften, Eberhard Karls Universität Tübingen, Wilhelmstr. 56, D-72074 Tübingen, Germany

*Corresponding author. Fachbereich Geowissenschaften, Eberhard Karls Universität Tübingen, Wilhelmstr. 56, D-72074 Tübingen, Germany. Telephone: +49-(0)7071-29-73155. Fax: +49-(0)7071-29-3060. E-mail: simon-johannes.braunger@student.uni-tuebingen.de

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ABSTRACT

The Miocene Kaiserstuhl Volcanic Complex (Southwest Germany) consists largely of tephritic to phonolitic rocks, accompanied by minor nephelinitic to limburgitic and melilititic to haüynitic lithologies associated with carbonatites. Based on whole-rock geochemistry, petrography, mineralogy and mineral chemistry, combined with mineral equilibrium calculations and fractional crystallization models using the Least Square Fitting Method, we suggest that the Kaiserstuhl was fed by at least two distinct magma sources. The most primitive rock type of the tephritic to phonolitic group is rare monchiquite (basanitic lamprophyre) evolving towards tephrite, phonolitic tephrite, phonolitic noseanite, nosean phonolite and tephritic phonolite by fractional crystallization of variable amounts of clinopyroxene, amphibole, olivine, spinel/magnetite, garnet, titanite, plagioclase and nosean. During this evolution, temperature and silica activity (a_{SiO_2}) decrease from about 1100°C and $a_{\text{SiO}_2} = 0.6\text{--}0.8$ to 880°C and $a_{\text{SiO}_2} = \sim 0.2$. At the same time, oxygen fugacity (f_{O_2}) increases from $\Delta\text{FMQ}^* = +2\text{--}3$ to $\Delta\text{FMQ}^* = +3\text{--}5$, with ΔFMQ^* being defined as the log f_{O_2} deviation from the silica activity-corrected FMQ buffer curve. Nephelinitic rocks probably derive by fractionation of mostly olivine, spinel/magnetite, melilite, perovskite and nepheline from an olivine melilititic magma. The nephelinitic rocks were formed at similarly high crystallization temperatures ($>1000^\circ\text{C}$) and evolve towards limburgite (hyalo-nepheline basanite) by an increase of silica activity from about $a_{\text{SiO}_2} = 0.4\text{--}0.5$ to $a_{\text{SiO}_2} = 0.5\text{--}0.9$, whilst redox conditions are buffered to ΔFMQ^* values of around +3. Häüyne melilitite and the more evolved (melilitite) haüynite may equally be derived from an olivine melilitite by more intense olivine and less melilite fractionation combined with the accumulation of haüyne, clinopyroxene and spinel. These rocks were crystallized at very low silica activities ($a_{\text{SiO}_2} \leq 0.2$) and highly oxidized conditions ($\Delta\text{FMQ}^* = +4\text{--}6$). Even higher oxygen fugacities ($\Delta\text{FMQ}^* = +6\text{--}7$) determined for the carbonatite suggests a close genetic relation between these two groups. The assemblage of carbonatites with highly oxidized silicate rocks is typical of many carbonatite occurrences worldwide, at least for those associated with melilititic to nephelinitic silicate rocks. Therefore, we suggest that the existence of highly oxidized carbonate-bearing sublithospheric mantle domains is an important prerequisite to form such complexes.

Key words: alkaline magmatism; carbonatites; fractional crystallization; Kaiserstuhl; mantle sources; redox conditions

INTRODUCTION

Most of the about 500 known carbonatite occurrences worldwide are spatially and temporally associated with alkaline rocks, which implies that both rock groups are genetically closely related (e.g. Woolley, 2001; Woolley & Kjarsgaard, 2008). Two major hypotheses for the formation of carbonatites are discussed: (1) partial melting of a carbonate-bearing peridotite leading to primary carbonatitic magmas (e.g. Wallace & Green, 1988; Harmer *et al.*, 1998; Ying *et al.*, 2004) and (2) fractional crystallization of a carbonate-bearing (nephelinitic to melilititic) parental silicate melt resulting in the formation of exsolved/residual carbonatitic magmas (Gittins, 1989; Kjarsgaard & Peterson, 1991; Gittins & Jago, 1998; Halama *et al.*, 2005; Brooker & Kjarsgaard, 2011).

The source region for most carbonatites and associated alkaline rocks is the (sub)lithospheric mantle (Bell & Simonetti, 2010). The oxidation state of the Earth's lithospheric mantle, however, varies over several orders of magnitude, which is commonly expressed as the log-deviation from the synthetic fayalite–magnetite–quartz (FMQ) buffer (e.g. Frost, 1991). For example, subduction-related peridotites record an average oxidation state of $\Delta\text{FMQ} > +0.5$, whereas cratonic mantle lithosphere shows a much more reduced average oxidation state of almost $\Delta\text{FMQ} = -3$ (e.g. Frost & McCammon, 2008; Foley, 2011 and references therein). Highly variable redox conditions (and silica activities) during the crystallization of alkaline rocks are reflected in their diverse mineralogy, as rocks with olivine–augite–Fe–Ti oxide assemblages record $f\text{O}_2$ values that are several log units lower than rocks with garnet–augite–titanite assemblages (e.g. Marks *et al.*, 2008a; Marks & Markl, 2017). Only few attempts have been undertaken to determine the redox conditions during the crystallization of carbonatites, probably due to the lack of suitable mineral assemblages. The few estimated $f\text{O}_2$ values all cluster around (Friel & Ulmer, 1974; Gaspar & Wyllie, 1983; Treiman & Essene, 1984; Ryabchikov *et al.*, 2008) or below the FMQ buffer (Milani *et al.*, 2016). The oxidation state of alkaline rocks associated with carbonatites has rarely been determined. In some cases, they are assumed to have been formed at $f\text{O}_2$ conditions around the FMQ buffer (Ryabchikov & Kogarko, 2006), whereas in other cases highly oxidized conditions (4 to 5 log units above FMQ) have been suggested (Halama *et al.*, 2005; Marks *et al.*, 2008a). This apparent discrepancy between the redox states of carbonatites and associated silicate rocks is noteworthy, since carbonatitic magma production might be triggered by hydrous redox (e.g. Taylor & Green, 1987), or carbonate redox melting (Foley, 2011). However, to date no attempt has been made to determine the redox conditions for carbonatites and associated alkaline silicate rocks at a given locality.

The Kaiserstuhl Volcanic Complex (KVC) in south western Germany is a well-suited locality for such investigations, as it consists of a variety of alkaline

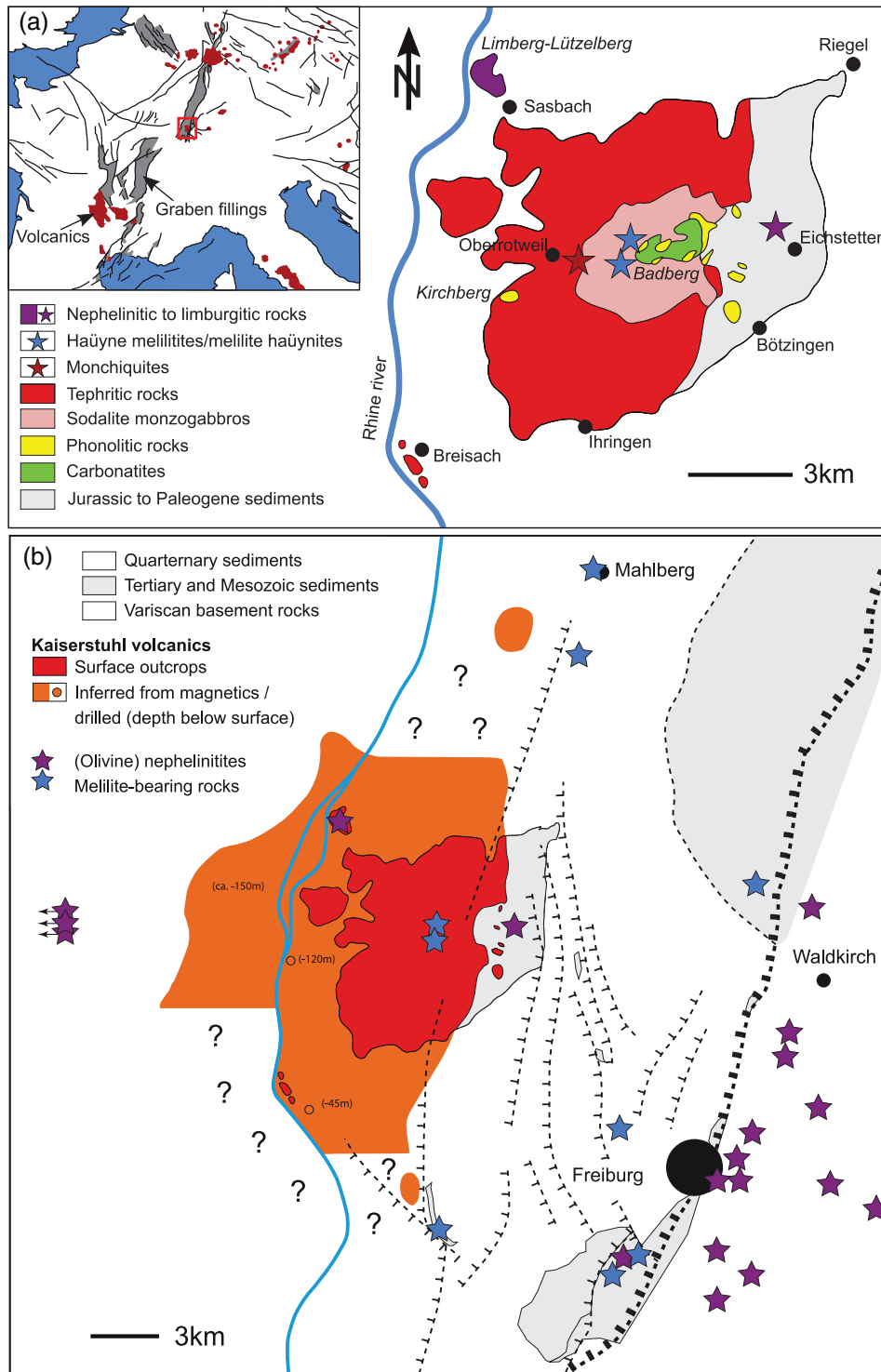
silicate rocks (primitive olivine-bearing tephritic/basanitic and nephelinitic rocks as well as evolved and garnet-bearing phonolites), which are closely associated with carbonatitic rocks (e.g. Keller *et al.*, 1990; Wimmenauer, 2003). However, despite being a classical carbonatite locality, published mineral and whole-rock compositional data for the KVC are fragmentary. No comprehensive petrological study of the KVC exists to date and the detailed genetic relationships between the various rock types are not fully understood (e.g. Schleicher *et al.*, 1990, 1991; Panina *et al.*, 2000). In order to shed light on the formation and evolution of the KVC and similar complexes in general, we present a detailed geochemical, mineralogical and mineral chemical characterization of the different rock types. Combined with fractional crystallization computations and thermodynamic estimates for the crystallization conditions (P , T , $a\text{SiO}_2$, $f\text{O}_2$), we show how these parameters influence the mineralogical and geochemical evolution of silicate magmas and how the various KVC rocks are genetically related to one another.

GEOLOGY

The KVC is part of the European Cenozoic Volcanic Province, situated in the southern part of the Upper Rhine Graben, about 20 km NW of Freiburg, southwest Germany (Fig. 1a; Wilson & Downes, 1991). The Upper Rhine Graben formed as a consequence of the stress field induced by the Alpine Orogeny, which caused lithospheric thinning below the KVC (Edel *et al.*, 1975; Ziegler, 1982). This enabled regional asthenospheric upwelling and the formation of the KVC, which was active between 19 and 16 Ma (e.g. Wimmenauer, 2003; Kraml *et al.*, 2006 and references therein).

The KVC was probably fed by several magma sources: a hypothetical K-basanitic magma produced large volumes of tephrites, sodalite monzogabbros, phonolitic tephrites and phonolites (Wimmenauer, 1962; Keller, 1984a). A volumetrically minor sodic rock series consists of olivine nephelinites, basanitic nephelinites and limburgites that are mainly present at the spatially associated Limberg-Lützelberg Complex near Sasbach (Fig. 1a). Melilite-bearing dyke rocks [häüyne melilitites and (melilite) häüynites] and intrusive as well as extrusive carbonatites are largely restricted to the central part of the KVC and are believed to be genetically related to the latter group (e.g. Keller, 1978, 1984a; Keller *et al.*, 1990; Schleicher *et al.*, 1990, 1991).

The original size of the Kaiserstuhl volcano during its active period is difficult to reconstruct because of: (i) intense deposition of Quaternary sediments in the Rhine valley and (ii) long-lasting tectonic activity in the Rhine graben during and after the formation of the KVC. Along the eastern flank of the KVC a considerable amount of erosion happened, probably due to westward tilting of parts of the area (Wimmenauer, 2003). Based on published and ongoing geophysical



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Fig. 1. Geology of the Kaiserstuhl Volcanic Complex. (a) Geological map of the European Cenozoic Volcanic Province and the Kaiserstuhl Volcanic Complex. Modified after Keller *et al.* (1990). (b) Sketch map showing the surface exposure of the Kaiserstuhl volcanic rocks combined with sub-surface information from drill holes (Groschopf *et al.* 1996) and geomagnetic work (Pucher, 1991; Bouiflane, 2008; terratec Geophysical Services, unpubl. data). Purple and blue stars indicate the known occurrences of nephelinitic and melilititic rocks, respectively (after Wimmenauer, 1952).

investigations (Pucher, 1991; Bouiflane, 2008; terratec Geophysical Services, unpublished data) and drill-core data (Groschopf *et al.*, 1996; Wimmenauer, 2003) in the immediate vicinity, large volumes of volcanic rocks are

hidden below the surface north, east and south of the KVC (Fig. 1b). Thus, the KVC as we see it today is no more than the diminished ruin of a once much larger volcanic complex that originally consisted of a group of

volcanic centres, comparable in size to, e.g. the Mt. Vulture Volcanic Complex (Italy). Further, nephelinitic and melilititic rocks occur in the immediate vicinity of the KVC (purple and blue stars on Fig. 1b), with the most prominent one being the olivine melilitite stock of Mahlberg (e.g. [Wimmenauer, 1952](#); [Dunworth & Wilson, 1998](#)). Together with the melilititic rocks of the well-known volcanic fields of Urach (about 150 km NE) and Hegau (about 100 km SE), such strongly SiO₂-undersaturated magmas are prominently present on a regional scale, at least partly contemporaneous with the volcanic activity at Kaiserstuhl.

SAMPLE LOCALITIES

In total, more than 150 samples from surface outcrops and drill holes at Orberg (KB2 1991; [Sigmund, 1996](#)) and Horberig (KB3 1991; [Blust, 1993](#)) were examined for this study (Table 1; Supplementary Data Fig. S1; supplementary data are available for downloading at <http://www.petrology.oxfordjournals.org>). Out of these, 89 were investigated in detail by means of whole-rock and/or mineral analyses.

Most samples were taken from the centre of the KVC, which shows the greatest diversity of rock types (carbonatites, melilite- and haüyne-bearing rocks, tephritic and phonolitic rocks). However, several samples from the Limberg-Lützelberg complex and the phonolite quarries at Kirchberg and Fohberg (NW of Bötzingen), as well as rare olivine tephrites (near Breisach), and tephritic and nephelinitic rocks in the western and eastern part of the KVC were also included.

PETROGRAPHY

The large variety of KVC rocks was classified following the scheme of the *International Union of Geological Sciences, Subcommittee on the Systematics of Igneous Rocks*, as summarized in [Le Maitre et al. \(2002\)](#), avoiding the (still frequently used) local names wherever possible (Table 2). We distinguish the following four rock groups:

Nephelinitic to limburgitic rocks

Olivine nephelinites and a basanitic nephelinite contain phenocrysts of olivine (partly altered to goethite and yellow clay minerals) and clinopyroxene (Fig. 2a), the latter additionally occurring as a subhedral to anhedral groundmass phase. Subhedral to anhedral spinel was only found as inclusions in olivine phenocrysts. Magnetite appears as inclusions in the rim areas of olivine and clinopyroxene phenocrysts and as a groundmass phase, where it occasionally shows a composite texture with ilmenite. Rarely, anhedral biotite appears in the nepheline-dominated matrix (Fig. 2a) that might also contain some subhedral plagioclase (HTAC1433).

Limburgites (hyalo-nepheline basanites) have a glassy groundmass and contain olivine (both as euhedral and anhedral grains that are partly altered to

iddingsite; Fig. 2b) and sector- or oscillatory zoned euhedral clinopyroxene phenocrysts (rarely with greenish cores). The latter also occurs as small groundmass needles. Subhedral to euhedral spinel appears as inclusions in olivine and as a groundmass phase, locally evolving towards magnetite (Fig. 2c). Magnetite also occurs as inclusions in clinopyroxene.

Olivine tephrites contain several mm in size, rounded macrocrysts of olivine (frequently altered to iddingsite) that may contain spinel inclusions, and euhedral, strongly altered olivine in the groundmass (Fig. 2d). Oscillatory-zoned rose-brownish clinopyroxene appears in variable sizes and commonly contains magnetite inclusions in its darker rim areas (Fig. 2e) as well as rare inclusions of olivine. Plagioclase forms subhedral grains in the groundmass and dominates over alkali feldspar.

Melilititic to haüynitic rocks

Haüyne melilitites, melilite haüynites and haüynites occur as porphyritic dyke rocks and contain variable amounts of melilite, haüyne, clinopyroxene, and magnetite phenocrysts (Fig. 2f). All of these phases also occur in their groundmass. In haüyne melilitites, perovskite appears as phenocrysts and as smaller cruciform-twinning crystals in the matrix (Fig. 2g). Melilite haüynites and haüynites, however, lack perovskite phenocrysts, but contain euhedral garnet instead (Fig. 2h). Haüyne is zoned with the cores containing exsolved Fe-sulfides. Further felsic mineral phases are nepheline and rare alkali feldspar, which occurs as small flakes and represents a late mineral phase. Apatite is ubiquitous. Notably, a large proportion of calcite is found in haüyne melilitites as individual anhedral grains (Fig. 2f), within cracks and as an alteration product of melilite.

Coarse-grained haüynolites and melteigites appear as cm- to dm-sized xenoliths in the foiditic and phonolitic rocks ([Bakhashwin, 1975](#); [Seeger, 2015](#)). Yellowish euhedral clinopyroxene occurs in variable modal amounts and sizes, typically with greenish rims (Fig. 3a). Oscillatory zoned garnet is present as euhedral to subhedral crystals, in places intergrown with clinopyroxene (Fig. 3b). Haüyne (commonly replaced by natrolite) is only present in the haüynolites, whereas nepheline occurs as an intercumulus phase in melteigites. The latter can additionally contain tabular plagioclase (Fig. 3c). Occasionally, magnetite, pyrite and pyrrhotite can be observed. Titanite and apatite are ubiquitous, while perovskite (Fig. 3d) is only present in some melteigites.

Tephritic to phonolitic rocks

This heterogeneous group of rocks constitutes the dominant part of the KVC (see details in [Kim, 1985](#)) and can be divided into primitive (monchiquites), intermediate (tephrites, phonolitic tephrites/sodalite monzogabbros, mesocratic tephritic phonolites and melanocratic phonolites)

Table 1: Location with UTM coordinates of the investigated samples.

Number	Sample	Location	UTM coordinates	Name
1	HTAC1446	Limberg	32 U 396072 5333966	Limburgite
2	HTAC1443	Limberg	32 U 396167 5333725	Limburgite
3	HTAC1438	Limberg	32 U 396187 5333554	Limburgite
4	HTAC1436	Ltzelberg	32 U 396526 5333258	Olivine nephelinite
5	HTAC1415	Katharinenberg	32 U 402288 5329644	Olivine sövite
6	HTAC1433	Eichstetten/ Schenkelgrub	32 U 404972 5328280	Basanitic nephelinite
7	HTAC1412	Orberg	32 U 402482 5329092	Phonolitic tephrite
7	HTAC1393	Orberg	32 U 402484 5329085	Phonolitic tephrite
8	HTAC1356	Orberg	32 U 402640 5328781	Monticellite sövite
8	HTAC0231	Orberg	32 U 402626 5329020	(Wollastonite-bearing) nosean phonolite
8	HTAC0232	Orberg	32 U 402665 5329041	Phonolitic tephrite
9	HTAC1291	Orberg Drilling KB2	32 U 402477 5328680	Phonolitic noseanite
10	HTAC290	Degenmatt	32 U 403049 5328638	Mesocratic tephritic phonolite
10	HTAC291	Degenmatt	32 U 403041 5328629	Phonolitic noseanite
10	HTAC293	Degenmatt	32 U 403038 5328620	(Wollastonite-bearing) nosean phonolite
11	HTAC299	Degenmatt	32 U 403001 5328410	Phonolitic noseanite
12	HTAC1354	Haselschacher Buck	32 U 402323 5328107	Olivine sövite
13	HTAC1391	Schelingen	32 U 401885 5328418	Sodalite monzogabbro
14	HTAC0243	Schelingen	32 U 401542 5328576	(Wollastonite-bearing) nosean phonolite
14	HTAC0244	Schelingen	32 U 401528 5328580	Phonolitic tephrite
14	HTAC1397	Schelingen	32 U 401660 5328612	Phonolitic tephrite
14	HTAC1386	Schelingen	32 U 401660 5328612	(Wollastonite-bearing) nosean phonolite
14	HTAC1407	Schelingen	32 U 401660 5328612	Phonolitic noseanite
15	HTAC1411	Oberbergen-Schelingen	32 U 401064 5328504	Phonolitic tephrite
16	HTAC1365	Hessleter Buck	32 U 400021 5328614	Hayne melilitite
16	HTAC1427	Pulverbuck	32 U 399970 5328595	Phonolitic tephrite/ sodalite monzogabbro
16	HTAC1406	Pulverbuck	32 U 399918 5328612	Phonolitic tephrite
17	HTAC1383	Pulverbuck	32 U 399856 5328950	Tephritic phonolite
17	HTAC1400	Pulverbuck	32 U 399842 5328960	Tephrite
18	HTAC1341	Mondhalde	32 U 399084 5328736	(Wollastonite-bearing) nosean phonolite
19	HTAC252	Horberig	32 U 400563 5327867	Phonolitic noseanite
19	HTAC1332, 1425	Horberig	32 U 400587 5327839	Hayne melilitite
19	HTAC1333	Horberig	32 U 400587 5327839	Melilite haynite
19	HTAC1336	Horberig	32 U 400563 5327837	Phonolitic noseanite
20	HTAC1387	Horberig	32 U 400636 5327773	Phonolitic noseanite
20	HTAC0218	Horberig	32 U 400801 5329041	Phonolitic tephrite (continued)
20	HTAC1374	Horberig	32 U 400729 5327814	(Wollastonite-bearing) nosean phonolite
20	HTAC1379	Horberig	32 U 400851 5327912	Phonolitic noseanite/ (Plagioclase-bearing) melteigite
20	HTAC1382	Horberig	32 U 400652 5327827	(Wollastonite-bearing) nosean phonolite
20	GM1430	Horberig	32 U 400652 5327810	Nosean phonolite/haynolite
20	PS2, 6-8, 10a	Horberig	32 U 400652 5327810	Nosean phonolite
20	PS3, 4	Horberig	32 U 400652 5327810	Nosean phonolite/haynolite
20	PS5	Horberig	32 U 400652 5327810	Nosean phonolite/melteigite
21	HTAC1110	Horberig Drilling KB3	32 U 400940 5327843	Melanocratic phonolite
21	HTAC1113, 1125, 1133, 1217	Horberig Drilling KB3	32 U 400940 5327843	Melilite haynite
21	HTAC1117	Horberig Drilling KB3	32 U 400940 5327843	Phonolitic tephrite
21	HTAC1120	Horberig Drilling KB3	32 U 400940 5327843	Melanocratic phonolite
21	HTAC1125, 1133, 1217	Horberig Drilling KB3	32 U 400940 5327843	Melilite haynite
21	HTAC1172	Horberig Drilling KB3	32 U 400940 5327843	Hayne melilitite
21	HTAC1241	Horberig Drilling KB3	32 U 400940 5327843	Mesocratic tephritic phonolite
22	HTAC213	Horberig	32 U 400849 5327828	Haynite
22	HTAC214	Horberig	32 U 400849 5327828	Phonolitic noseanite/ (Wollastonite-bearing) nosean phonolite
22	HTAC216	Horberig	32 U 400796 5327784	Phonolitic noseanite
22	HTAC217	Horberig	32 U 400729 5327814	Phonolitic noseanite
23	HTAC201	Badloch	32 U 401181 5327456	Mesocratic tephritic phonolite

(continued)

Table 1: Continued

Number	Sample	Location	UTM coordinates	Name
23	HTAC202	Badloch	32 U 401242 5327438	Phonolitic noseanite
24	HTAC267	East of Badloch	32 U 401359 5327451	Mesocratic tephritic phonolite
24	HTAC1360	East of Badloch	32 U 401366 5327465	Haynite
24	HTAC1390	East of Badloch	32 U 401501 5327470	(Wollastonite-bearing) nosean phonolite
24	HTAC0268	East of Badloch	32 U 401856 5327550	Nosean syenite
25	HTAC1398	Endhalde	32 U 403779 5326955	(Wollastonite-bearing) nosean phonolite
26	HTAC1376, 1377, 1378	Fohberg	32 U 403676 5325793	(Wollastonite-bearing) nosean phonolite/Haynolite
27	HTAC1389	Fohberg	32 U 403881 5325801	(Wollastonite-bearing) nosean phonolite
28	HTAC1345	Fohberg	32 U 403655 5325623	(Wollastonite-bearing) nosean phonolite
29	HTAC1344	Totenkopf	32 U 399812 5327020	Sodalite monzogabbro
30	HTAC1408	Totenkopf	32 U 399669 5327099	Phonolitic tephrite
31	HTAC1396, 1402	Oberrotweil	32 U 398380 5327327	Monchiquite
32	HTAC1369	Henkenberg Drilling II	32 U 396571 5327640	Monchiquite
33	HTAC1343	Sponeck	32 U 394700 5329987	Sodalite monzogabbro
34	HTAC1351, 1392	Kirchberg	32 U 396728 5326421	Tephritic phonolite
34	HTAC1375	Kirchberg	32 U 396728 5326421	(Wollastonite-bearing) nosean phonolite
35	HTAC1424	Bchsenberg	32 U 396249 5325222	Tephrite
36	HTAC1423	Ihringen-Breisach	32 U 397401 5321607	Olivine tephrite
37	HTAC1434	Breisach	32 U 393825 5320779	Olivine tephrite

Numbers refer to [Supplementary Data Fig. S1](#).

Table 2: Average modal mineralogy of investigated Kaiserstuhl rocks

	Olivine	Clinopyroxene	Spinel/magnetite	Garnet	Amphibole/mica	Titanite	Perovskite	Plagioclase	Alkali feldspar	SGM ¹	Nepheline	Melilite	Wollastonite	Apatite	Carbonate	Colour code	Old (local) name
Nephelinitic to limburgitic rocks																	
Olivine nephelinite	22	42	8		2						26						Olivine nephelinite
Basanitic nephelinite	18	46	8		2			6			20						Olivine nephelinite
Limburgite	16	46	10					X			X						Limburgite
Olivine tephrite	8	42	8					32	4	6				Tr			Olivine tephrite
Melilitic to haüynitic rocks																	
Haüyne melilitite	6	8			6						26	8	34	Tr	8		Bergalite
Melilitite haüynite	20	8			2		4				8		8	Tr			Haüynophyre
Haüynite	24	6	16	18		2					36	8	8	Tr			Haüynophyre
Haüynolite	6	2	10			Tr	Tr		4		40	8		Tr			
(Plagioclase-bearing) melteigite	70	8	2	4		Tr	Tr	6			82			Tr			
Melteigite	76	6	6			Tr	Tr				12			Tr			
Tephritic to phonolitic rocks																	
Monchiquite	10	44	10		4			X			X						Olivine monchiquite
Tephrite	38	10			2			32	6	12				Tr			Tephrite
Phonolitic tephrite/sodalite monzogabbro	36	8			4	Tr		32	12	8				Tr			Phonolitic tephrite/essexite
Mesocratic tephritic phonolite	30	14			10			10	24	12				Tr			Gaiteite
Melanocratic phonolite	40	8			8				22	22				Tr			Shonkinite porphyry
Phonolitic noseanite	18				14				20	36	8		4	Tr			Tinguaite
Nosean phonolite/syenite	14				10		Tr		48	28				Tr			Phonolite/ledmorite-foid syenite
(Wollastonite-bearing) nosean phonolite	6				2				68	14			10	Tr			Phonolite
Tephritic phonolite	6	Tr	2			Tr		10	76	6				Tr			Phonolite
Carbonatites																	
Olivine sövite	10		8											Tr	82		Magnetite-Forsterite-Sövite
Monticellite sövite	14 ²		4		Tr									2	80		Magnetite-Monticellite-Sövite

(X) Minerals are present in the glass matrix. (Tr) Trace minerals. ⁽¹⁾ SGM, Sodalite group Minerals; ⁽²⁾ Monticellite instead of olivine.

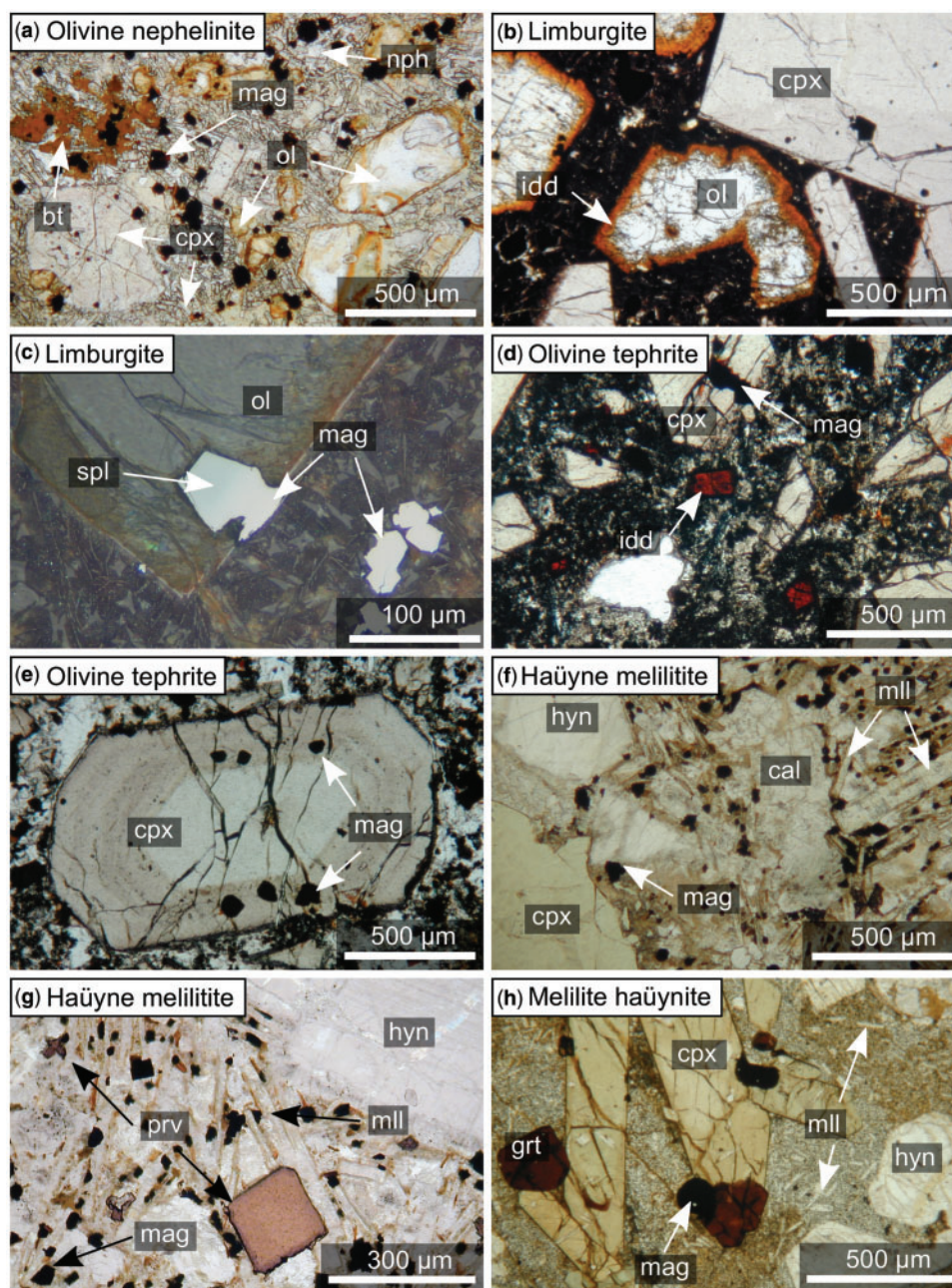


Fig. 2. Mineral textures in nephelinitic to limburgitic and melilititic to h aüynitic rocks. (a) Olivine (ol) and clinopyroxene (cpx) phenocrysts in olivine nephelinite with nepheline (nph), biotite (bt) and magnetite (mag) being typical groundmass phases (HTAC1436–2). (b) Clinopyroxene and olivine macrocrysts set in a largely glassy groundmass in limburgite HTAC1446. Red olivine rim consists of iddingsite (idd). (c) Combined transmitted and reflected light image of limburgite HTAC1446. A spinel (spl) inclusion in olivine gradually evolves into magnetite. (d) In olivine tephrites, groundmass olivines are totally replaced by iddingsite (HTAC 1423–2). (e) Magnetite inclusions in zoned clinopyroxene (olivine tephrite HTAC1423–1). (f) The main mineral phases of h aüyne melilitites are melilitite (mll), magnetite (mag), clinopyroxene and h aüyne (hyn; HTAC1365). (g) Perovskite (prv) phenocrysts and cruciform-twinned groundmass perovskite in h aüyne melilitite (HTAC1332). (h) Melilitite h aüynite (HTAC1133) containing phenocrysts of garnet (grt), clinopyroxene, magnetite and h aüyne.

and evolved members (phonolitic noseanites, nosean phonolites/nosean syenites, and tephritic phonolites), with systematically changing mineralogical features.

Monchiquites contain olivine and clinopyroxene phenocrysts (rarely with greenish cores); the latter mineral also occurs as a groundmass phase. In one sample (HTAC1396a), needle-shaped olivine was observed embedded in zeolite minerals as an inclusion in

clinopyroxene. Spinel generally occurs as inclusions in olivine, whereas magnetite is mostly present in clinopyroxene and in the matrix. If present in the groundmass, spinel is always rimmed by magnetite (Fig. 4a). The groundmass is largely glassy.

Tephrites are porphyritic and contain euhedral yellowish pale green and oscillatory zoned phenocrysts of clinopyroxene (rarely with greenish cores), which is

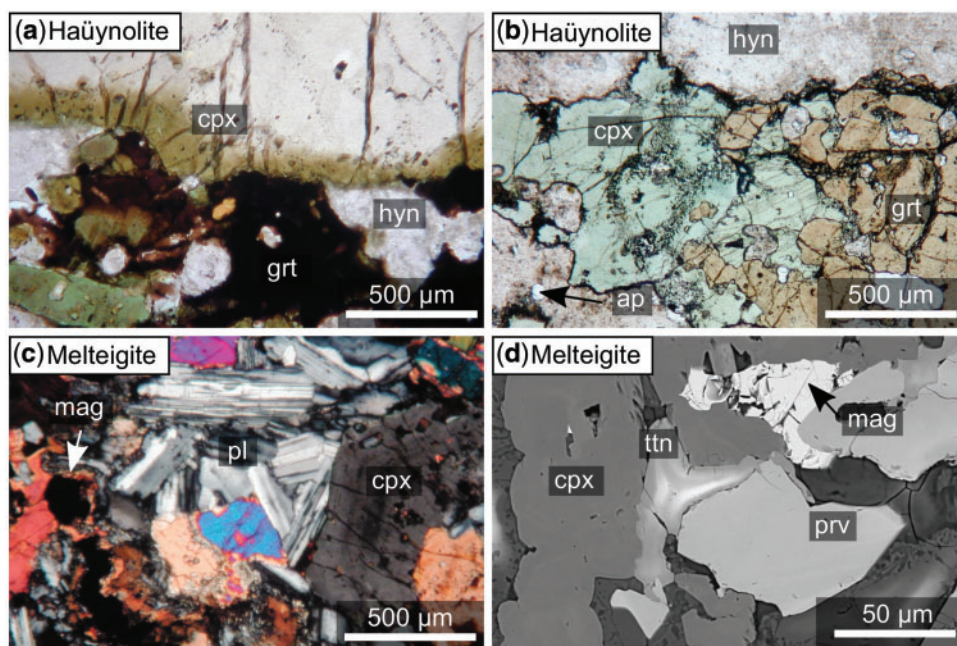


Fig. 3. Mineral textures in haüynolites and melteigites. (a) Garnet co-precipitated with dark green clinopyroxene (haüynolite GM1430C). (b) Intergrowth of clinopyroxene and garnet in haüynolite HTAC1376, with haüyne being altered to zeolites. (c) Plagioclase, clinopyroxene and magnetite assemblage in melteigite HTAC1379. (d) Perovskite overgrown by titanite in melteigite HTAC1379.

also present as a groundmass phase. Magnetite mainly appears in the matrix, but occurs as inclusions in marginal zones of pyroxene or is partially surrounded by clinopyroxene (Fig. 4b). The amount, type and size of feldspar grains varies strongly among individual samples. Plagioclase occurs as a groundmass phase and is partly altered to zeolites. Alkali feldspar occasionally surrounds plagioclase, but also appears as individual grains. Contrary to rarely occurring leucite crystals, which are always completely replaced by analcime, euhedral sodalite was only partially affected by this late stage alteration. Some of the more evolved and hence alkali feldspar-bearing samples contain up to 1 mm biotite and amphibole phenocrysts (Fig. 4c). Biotite additionally occurs as small grains in the groundmass, and apatite is the most common accessory mineral. Sodalite monzogabbros are mineralogically very similar to tephrites but have an equigranular and medium-grained texture. Amphibole always occurs as subhedral to euhedral crystals (Fig. 4c), whereas biotite is generally anhedral (Fig. 4d). Both minerals partly replace oscillatory zoned clinopyroxene and mostly subhedral magnetite. The latter additionally occurs as inclusions in clinopyroxene. One sample (HTAC1344) contains composite ilmenite–magnetite intergrowths. Alkali feldspar may enclose tabular plagioclase, but also occurs as euhedral crystals. Apatite and titanite are common accessory phases.

Most phonolitic rocks (phonolitic noseanites, nosean phonolites) have an alkali feldspar-dominated groundmass and lack plagioclase, magnetite and olivine, but contain variable amounts of garnet and wollastonite, except for mesocratic tephritic phonolites and a melanocratic phonolite, which are mineralogically very similar

to the above-described tephritic rocks. Phonolitic noseanites (Fig. 4e) contain phenocrysts of bright green and locally needle-shaped clinopyroxene, brown garnet, nosean, nepheline and strongly altered wollastonite. Porphyritic nosean phonolites comprise euhedral bright green clinopyroxene I and variable amounts of subsequently crystallized dark green clinopyroxene II (Fig. 4f). Euhedral and oscillatory zoned brown garnet (Fig. 4g) is in places replaced by a darker garnet generation, which is associated with clinopyroxene II (Fig. 4f). Few samples contain magnetite and titanite, in one sample (HTAC1345) titanite replaces garnet. Magnetite- and titanite-free samples, however, contain euhedral wollastonite, often replaced by natrolite. Euhedral nosean is commonly replaced by analcime, natrolite or calcite. Oscillatory zoned alkali feldspar can be present as tabular phenocrysts and as smaller grains in the groundmass. Coarse-grained nosean syenites are rare and were only found in drill cores south of Badberg (Bakhashwin, 1975) and as xenoliths within the phonolitic rocks (Czygan, 1977). They contain tabular alkali feldspar, nosean (commonly replaced by natrolite), euhedral and greenish clinopyroxene (in some cases with a yellowish core), euhedral to subhedral oscillatory zoned garnet and blocky apatite. Magnetite is only present in one sample (HTAC268B). Titanite is ubiquitous, generally subhedral to euhedral. Only in one sample (HTAC1375) it is absent and wöhlerite ($\text{Na}_2\text{Ca}_4\text{ZrNb}[\text{F}|\text{O}_3](\text{Si}_2\text{O}_7)_2$) is present instead.

Tephritic phonolites (Kirchberg and Oberrotweil localities) are distinguished from nosean phonolites and phonolitic noseanites, as they contain anhedral garnet, show small amounts of nosean, and contain

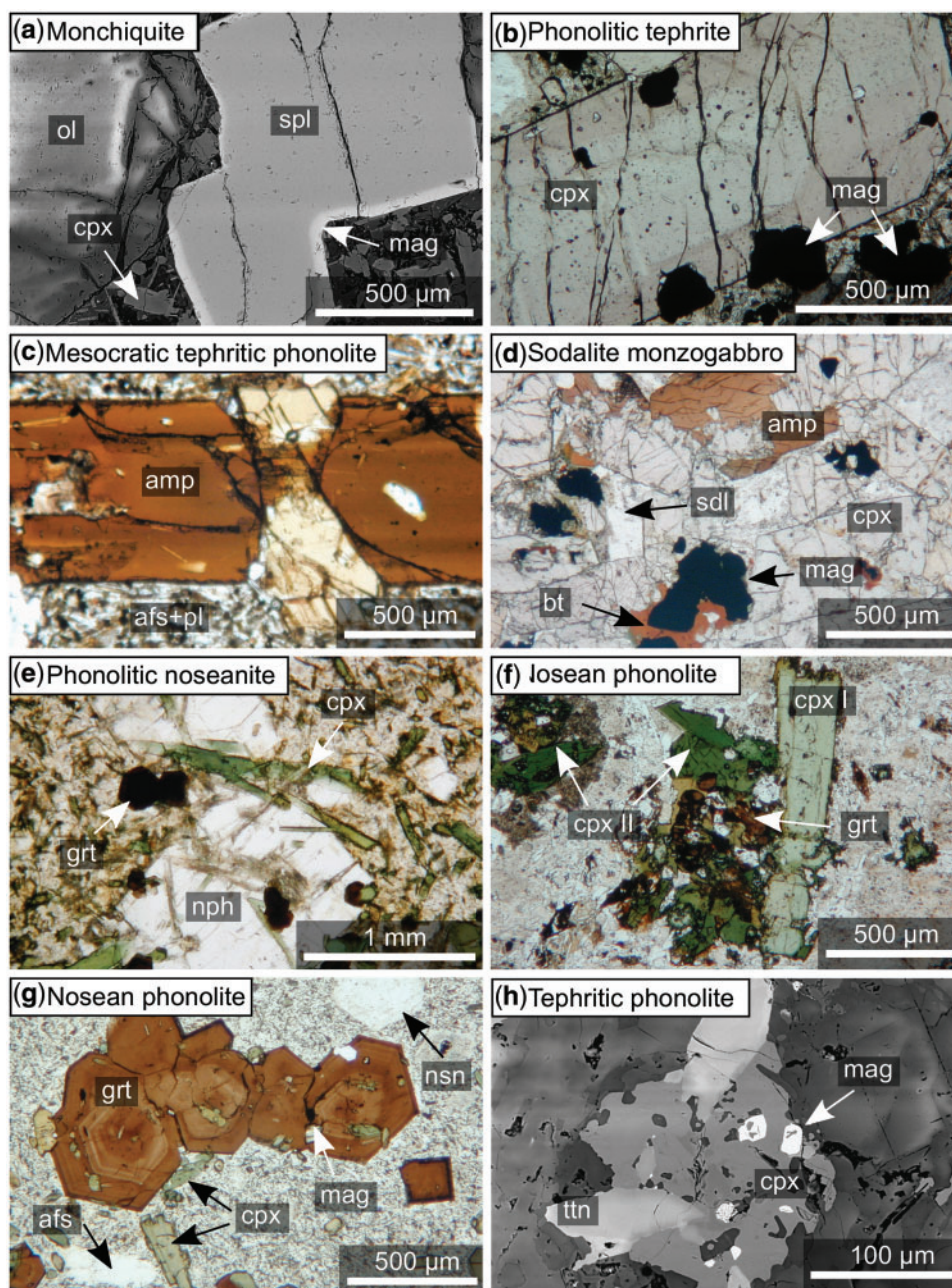


Fig. 4. Mineral textures in tephritic to phonolitic rocks. (a) Spinel rimmed by magnetite in monchiquite HTAC1396a (BSE-image). (b) Magnetite aligned in dark clinopyroxene rims (phonolitic tephrite HTAC1411b). (c) Amphibole phenocrysts (amp) embedded in fine-grained plagioclase and alkali feldspar in mesocratic tephritic phonolite HTAC1241. (d) Typical texture of a sodalite monzogabbro with amphibole, clinopyroxene, sodalite (sdl), minor amounts of biotite and magnetite (HTAC1391). (e) Phonolitic noseanites contain nepheline phenocrysts (HTAC216). (f) Two clinopyroxene generations in nosean phonolites (HTAC1345). Late-stage garnet is always associated with clinopyroxene II. (g) Zoned garnet associated with clinopyroxene, magnetite, nosean (nsn) and alkali feldspar in nosean phonolite PS8. (h) Tephritic phonolites (HTAC1351) contain the coexisting mineral assemblage of clinopyroxene, titanite and magnetite (BSE image).

plagioclase phenocrysts. Texturally, clinopyroxene, titanite and magnetite coexist with each other in these rocks (Fig. 4h). One nosean phonolite (HTAC1375) is transitional to tephritic phonolite as it shows the same mineral characteristics except for the absence of plagioclase crystals. Rare trachytes occur southwest of the Badberg (Sutherland, 1967; Koberski, 1992) and were reported from drill hole KB2 (Sigmund, 1996). They are

characterized by alkali feldspar phenocrysts, the absence of garnet and minor amounts of feldspathoid minerals.

Carbonatites

Carbonatites of the KVC include coarse-grained calcite-carbonatite (sövites), medium- to fine-grained calcite-

carbonatite (alvikites), rare medium- to fine-grained dolomite-carbonatite (beforsites) and minor extrusive rocks (lapillistones, tuffs, lavas) and have been described elsewhere in great detail (e.g. Keller, 1978, 1981; Lehnert, 1989). Three sövites from Orberg, Haselschacher Buck and Katharinenberg have been chosen for this study. In addition to calcite, they contain magnetite, phlogopite, pyrochlore and apatite along with rare monticellite (Orberg; Fig. 5a) or olivine (Fig. 5b).

METHODS

Whole-rock analyses

Major elements were analysed at the Department of Geosciences, University of Tübingen (Germany), with a wavelength dispersive Bruker AXS S4 Pioneer XRF device (Rh-tube at 4 kW) with 32 standardised samples (compiled in Govindaraju, 1989). Prior to preparation, the samples were ground with an agate mill for 10 minutes. For the fused beads 1.5 g of dried sample powder (at 105 °C) was mixed with 7.5 g MERCK spectromelt A12 (mixture of 66 % Li-tetraborate and 34 % Li-metaborate) and melted at 1200 °C beads using an Oxiflux system from CBR analytical service. Detection limits and analytical errors are sample specific and usually in the range of 0.1 % (relative).

Trace elements were determined by ICP-MS (X-series 2, Thermo Fisher Scientific) at the Institute for Applied Geosciences, Karlsruhe Institute of Technology (Germany) after HNO₃-HF-HClO₄ acid digestions of powdered material (100 mg). To ensure a complete silicate decomposition, 40% HF (suprapur), 65% HClO₄ (normatom) and the pre-oxidized (65% HNO₃, subboiled) sample were heated in a closed Teflon vessel for 16 h at 120 °C. After evaporating the acids to incipient dryness, the residue was re-dissolved in 65% HNO₃ (subboiled) and evaporated again (three times) for purification purposes. The final residue was dissolved in 50 ml of ultrapure water. The quality assurance of the ICP-MS measurement of the trace elements was done by including the certified reference material CRM-TMDW-A (High-Purity standards, Inc.) into the protocol (accuracy: ± 10% for most elements). Mathematical corrections of Eu, Gd, Tb and Er were necessary due to high concentrations of Ba and light REE; ^{151/153}Eu was corrected for BaO interferences, ¹⁵⁷Gd for PrO and ¹⁵⁸Gd for CeO interferences, ¹⁵⁹Tb and ¹⁶⁷Er for NdO interferences. The mean value of both isotopes is given for Eu and Gd. To assure the quality of the whole procedure, six blanks and two certified reference materials (SY-2, SY-3; Govindaraju, 1994) were included in the digestion (accuracy: mainly ± 10%). The reproducibility (± 5% for most elements) was checked by digesting a phonolitic sample in triplicate.

Mineral analyses

Mineral analyses were performed using a JEOL JXA 8900 electron microprobe in wave-length dispersive

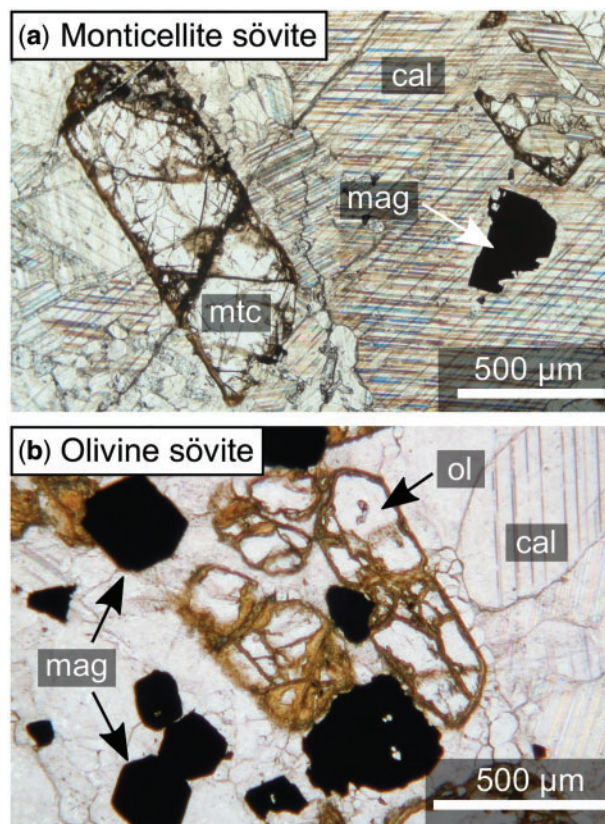


Fig. 5. Mineral textures in carbonatites. (a) Sövite carbonatite with monticellite (mtc) and magnetite macrocrysts (HTAC1356). (b) Sövite with olivine and magnetite phenocrysts (HTAC1354).

mode at the Department of Geosciences, University of Tübingen (Germany). Depending on the mineral, an acceleration voltage of 15 kV (felsic minerals) or 20 kV was chosen. The beam current was set to 20 nA, except for garnet, titanite and perovskite analyses which were acquired using a current of 12 nA. The peak counting times for major elements were 16 s and between 30 and 60 s for minor elements. Background counting times were half as long as the peak counting. For calibration, natural and synthetic standards were used. Peak overlap corrections involving Ba–Ti, V–Ti, F–Fe, Sm–Ce and Pr–La were implemented as well as an internal ZAF (garnet, titanite and perovskite) or $\phi\rho z$ correction of the raw data (Armstrong, 1991). Garnet, titanite and perovskite analyses were performed with a 5 µm beam diameter since test measurements executed with a smaller beam size led to fluctuating probe currents and visible destruction of the grains due to the long measurement time. Amphibole and biotite were measured with a 2 µm beam spot. Sodium diffusion in felsic minerals was avoided by measuring Na first and widening the beam diameter to 10 µm.

Quantification of intrinsic parameters

In order to constrain the crystallization conditions (P , T , fO_2 , $aSiO_2$) for the various rock types, several

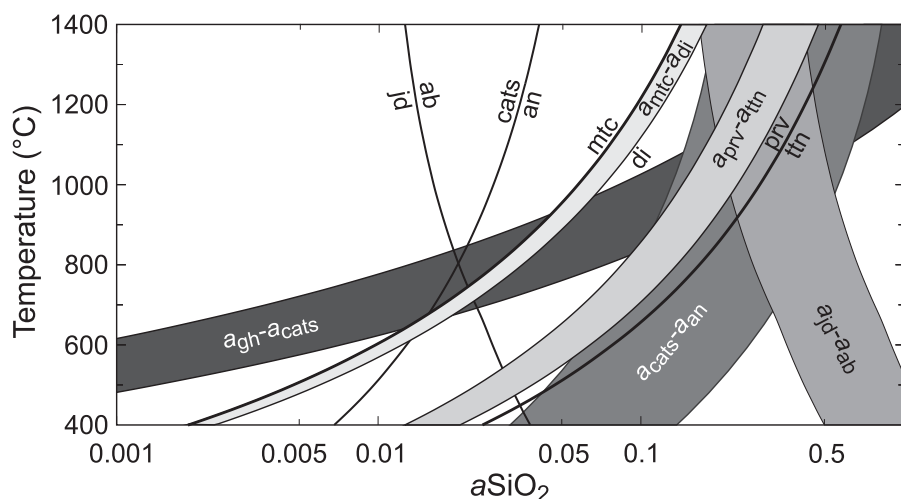


Fig. 6. Temperature- $a\text{SiO}_2$ plot illustrating the temperature dependence of the reactions used to determine silica activity. Thick lines represent reactions at unit activities, while the broad fields show their position when typical endmember activities are used.

geothermobarometers were applied and various activity-corrected phase diagrams were computed. Phase diagrams were computed and plotted using the *Perple_X* software package of Connolly (1990), implemented with the thermodynamic database of Holland & Powell (1998). Thermodynamic properties of titanite and perovskite were taken from Robie & Hemingway (1995) and were checked for internal consistency. Activities for endmember components of alkali feldspar, plagioclase, pyroxene, garnet, Fe–Ti oxides and olivine group minerals were obtained using the solution models of Holland & Powell (2003), Holland (1990), Wood (1979), Cosca *et al.* (1986) and Andersen *et al.* (1993), respectively. Nepheline, melilite, perovskite and titanite activities were calculated using a mixing-on-site model and by assuming oxygen substitution in titanite only being possible in one O site (Tropper & Manning, 2008). Based on the analyses of Weisenberger *et al.* (2014), wollastonite was treated as a pure phase.

Pressure estimations for some of the rocks were possible by applying the temperature-dependent Al-in hornblende barometer of Anderson & Smith (1995). Temperatures have been estimated using the feldspar solution model of Fuhrman & Lindsley (1988) and the amphibole-plagioclase thermometer of Holland & Blundy (1994). For applying the clinopyroxene-liquid thermometer of Putirka *et al.* (1996), we used the whole-rock data as a proxy for the liquid composition. Oxygen fugacity was calculated using the recently calibrated oxybarometer of Arató & Audétat (2017), which is based on the FeO and TiO₂ contents of magnetite and the melt. Similar to the clinopyroxene-liquid thermometer, calculations were only performed for samples where xenoliths, xenocrysts or mineral accumulations could be excluded. Combined $f\text{O}_2$, $a\text{SiO}_2$ and temperature estimations for some rocks were possible with the QUILF software of Andersen *et al.* (1993). For other rock types, we used combinations of the following silica-sensitive (reactions 1–7; Fig. 6) and redox-dependent phase equilibria (reactions 8–15):

1. $\text{NaAlSiO}_4 + \text{SiO}_2 = \text{NaAlSi}_2\text{O}_6$
2. $\text{NaAlSiO}_4 + 2 \text{SiO}_2 = \text{NaAlSi}_3\text{O}_8$
3. $\text{NaAlSi}_2\text{O}_6 + \text{SiO}_2 = \text{NaAlSi}_3\text{O}_8$
4. $\text{CaTiO}_3 + \text{SiO}_2 = \text{CaTiSiO}_5$
5. $\text{CaAl}_2\text{SiO}_6 + \text{SiO}_2 = \text{CaAl}_2\text{Si}_2\text{O}_8$
6. $\text{CaMgSiO}_4 + \text{SiO}_2 = \text{CaMgSi}_2\text{O}_6$
7. $2 \text{Ca}_2\text{Al}_2\text{SiO}_7 + 2 \text{SiO}_2 = \text{CaAlAlSiO}_6 + \text{Ca}_3\text{Al}_2\text{Si}_3\text{O}_{12}$
8. $9 \text{CaFeSi}_2\text{O}_6 + 2 \text{O}_2 = 3 \text{Ca}_3\text{Fe}_2\text{Si}_3\text{O}_{12} + \text{Fe}_3\text{O}_4 + 9 \text{SiO}_2$
9. $2 \text{CaSiO}_3 + 4 \text{CaFeSi}_2\text{O}_6 + \text{O}_2 = 2 \text{Ca}_3\text{Fe}_2\text{Si}_3\text{O}_{12} + 4 \text{SiO}_2$
10. $16 \text{CaFeSi}_2\text{O}_6 + 2 \text{CaTiSiO}_5 + 3 \text{O}_2 = 2 \text{Fe}_2\text{TiO}_4 + 6 \text{Ca}_3\text{Fe}_2\text{Si}_3\text{O}_{12} + 16 \text{SiO}_2$
11. $16 \text{CaFeSi}_2\text{O}_6 + 2 \text{CaTiO}_3 + 3 \text{O}_2 = 2 \text{Fe}_2\text{TiO}_4 + 6 \text{Ca}_3\text{Fe}_2\text{Si}_3\text{O}_{12} + 14 \text{SiO}_2$
12. $3 \text{CaFeSi}_2\text{O}_6 + 3 \text{FeTiO}_3 + \text{O}_2 = 3 \text{CaTiSiO}_5 + 2 \text{Fe}_3\text{O}_4 + 3 \text{SiO}_2$
13. $2 \text{CaFeSi}_2\text{O}_6 + 2 \text{Fe}_2\text{TiO}_4 + \text{O}_2 = 2 \text{CaTiO}_3 + 2 \text{Fe}_3\text{O}_4 + 4 \text{SiO}_2$
14. $2 \text{CaTiSiO}_5 + 2 \text{Fe}_3\text{O}_4 + 2 \text{SiO}_2 = 2 \text{CaFeSi}_2\text{O}_6 + 2 \text{Fe}_2\text{TiO}_4 + \text{O}_2$
15. $3 \text{Fe}_2\text{SiO}_4 + \text{O}_2 = 2 \text{Fe}_3\text{O}_4 + 3 \text{SiO}_2$

Relative oxygen fugacity (ΔFMQ) is generally expressed as the deviation from the temperature-dependent FMQ buffer curve in $\log f\text{O}_2$ units. However, the use of ΔFMQ is not consistent in the literature. Importantly, this buffer curve is also strongly $a\text{SiO}_2$ -dependent and may shift by several log units. Therefore, it is necessary to state $a\text{SiO}_2$ values along with relative oxygen fugacity estimates (e.g. Marks *et al.*, 2008a; Zaitsev *et al.*, 2012). We present our results relative to $a\text{SiO}_2$ -corrected FMQ buffer curves, expressed as ΔFMQ^* .

Fractional crystallization calculations

Fractional crystallization was modelled using a least squares fitting method based on the assumption that

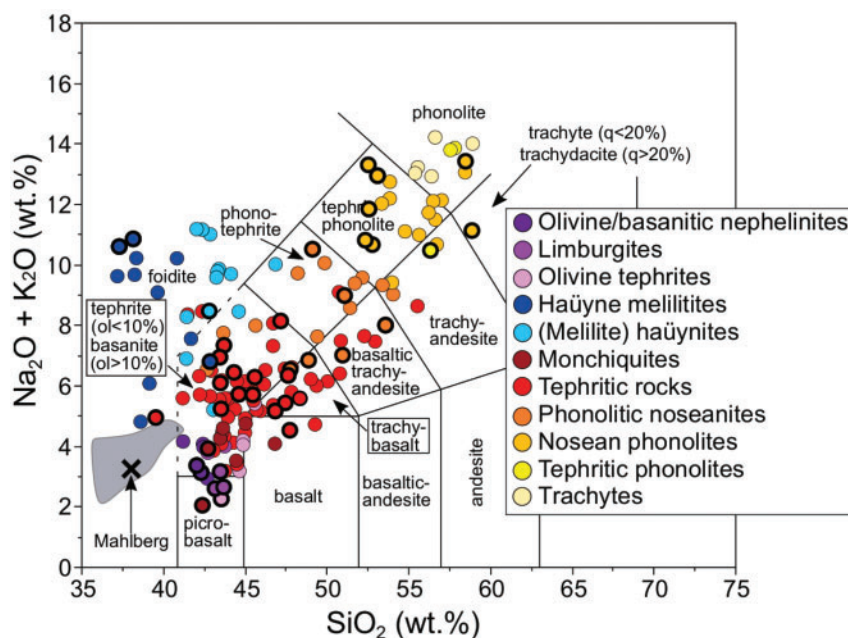


Fig. 7. Total alkalis ($\text{Na}_2\text{O}+\text{K}_2\text{O}$) vs SiO_2 for the silicate rocks of the KVC. New analyses are shown with thick rims, literature data from Bakhshwin (1975), Czygan (1977), Kim (1985), Keller *et al.* (1990), Blust (1993), Panina *et al.* (2000), Sigmund (1996), Wimmenauer (2003) and Grapes & Keller (2010). Grey area illustrates the compositional range of olivine melilitites from SW Germany (Keller *et al.*, 1990; Wilson *et al.*, 1995). Note that the loss of ignition of häüyne melilitites and olivine melilitites may reach ~ 10 wt %, which is not taken into account in this diagram, as all data are normalized to 100% (anhydrous), ignoring volatile components.

phenocrysts represent the principal products of that process. Whole-rock analyses were used as a proxy for the initial melt compositions and compositional data for the various mineral phases were used to compute the target melt. The quality of the outcome was evaluated by the residual sum of squares ($\text{SSQ}=[\text{SiO}_2^{\text{target}}-\text{SiO}_2^{\text{calculated}}]^2+[\text{TiO}_2^{\text{target}}-\text{TiO}_2^{\text{calculated}}]^2+\dots$).

RESULTS

Whole-rock data

New whole-rock analyses (Supplementary Data Table S1) together with literature data (Bakhshwin, 1975; Czygan, 1977; Kim, 1985; Keller *et al.*, 1990; Blust, 1993; Sigmund, 1996; Panina *et al.*, 2000; Wimmenauer, 2003; Grapes & Keller, 2010) provide a comprehensive overview of the compositional variation of the various lithologies at the KVC (Figs 7–9). The classic TAS diagram (Fig. 7) illustrates an increase of ($\text{Na}_2\text{O}+\text{K}_2\text{O}$) and SiO_2 concentrations from primitive olivine melilitites, which do not occur at the KVC but are present at several localities in SW Germany (Mahlberg, Urach, Hegau), through olivine/basanitic nephelinites, limburgites, olivine tephrites and monchiquites to tephritic and phonolitic rocks. Häüyne melilitites, melilite häüynites, and häüynites, however, form a distinct group with highly elevated alkali contents.

Two rock series can be distinguished based on the molar Na/K ratio (Fig. 8). The monchiquite is the most primitive rock of the low Na/K group ($< \sim 2$) and shows the highest contents of MgO. With decreasing MgO concentration, major elements either increase

($\text{Na}_2\text{O}+\text{K}_2\text{O}$, Al_2O_3) or decrease (CaO , Fe_2O_3). The correlation between MgO and Fe_2O_3 is typical of oxidized rock suites (Giehl *et al.*, 2013; Marks & Markl, 2017). The concentration of TiO_2 deviates slightly from this trend as the tephritic rocks show higher values than the monchiquites. The second group shows Na/K ratios ≥ 2 . From olivine melilitites via olivine/basanitic nephelinites, towards limburgites and olivine tephrites (with decreasing MgO), Al_2O_3 steadily increases, whereas $\text{Na}_2\text{O}+\text{K}_2\text{O}$, CaO, Fe_2O_3 and TiO_2 remain almost constant. Häüyne melilitites and (melilite) häüynites deviate from this pattern by higher $\text{Na}_2\text{O}+\text{K}_2\text{O}$, CaO and Al_2O_3 and lower Fe_2O_3 and TiO_2 concentrations, which additionally either positively (CaO, Fe_2O_3 , TiO_2) or negatively ($\text{Na}_2\text{O}+\text{K}_2\text{O}$, Al_2O_3) correlate with MgO.

With decreasing MgO, compatible elements such as Ni (up to 560 ppm) or Cr (up to 1400 ppm) steadily decrease, whereas incompatible elements such as Zr (≤ 770 ppm), Nb (≤ 620 ppm), La (≤ 350 ppm), Ce (≤ 630 ppm) and Pb (≤ 35 ppm) increase (Fig. 9). However, REE and HFSE concentrations in the häüyne melilitites and (melilite) häüynites are similar or even higher than those of the phonolitic rocks, which have lower MgO.

Primitive-mantle normalized (Palme & O'Neill, 2003) REE patterns of whole-rocks (Fig. 10) show a decrease from light to heavy REE. The normalized REE concentrations increase in parallel from monchiquite via tephrite to phonolitic tephrite. Phonolitic noseanites and especially nosean phonolites have steeper patterns and flatten out towards the HREE. The REE pattern of the

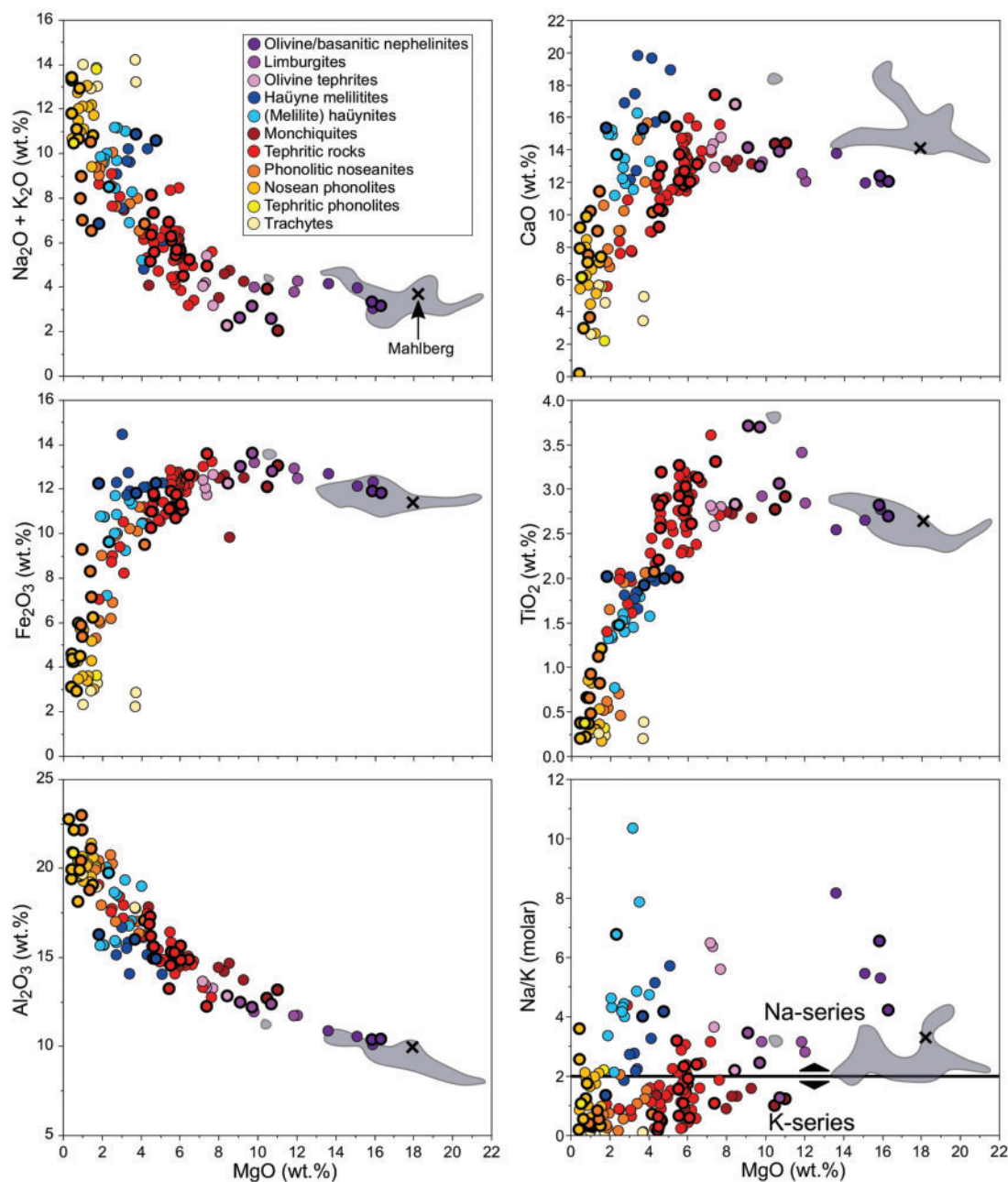


Fig. 8. Correlation diagrams for the various volcanic rocks of the KVC. Olivine melilitites from SW Germany are shown as a grey field (data from Keller *et al.*, 1990; Wilson *et al.*, 1995).

tephritic phonolite is less steep and increases from Ho to Lu. The nephelinitic to limburgitic rocks show a similar pattern to the monchiquites, whereas the melilititic to häüynitic rocks exhibit elevated concentrations and are depressed in the middle REE.

Mineral data

Olivine group

Olivine in the KVC is invariably Mg-rich (Fig. 11a; Supplementary Data Table S2), with Fo_{96-90} in carbonatites, Fo_{90-79} in olivine nephelinites (due to growth zonation), Fo_{86-79} in limburgites and Fo_{86-77} in monchiquites. Rounded macrocrysts in olivine tephrites are Fo_{85-83} ,

whereas groundmass olivines are Fo_{72-71} (including data from Kim, 1985, shown as rectangles in Fig. 11a). The larnite component is highest in olivine/basanitic nephelinites and monchiquites (up to 0.9 mol %) among the silicate rocks, whereas in some carbonatite samples, the larnite component exceeds 1 mol %. The Ni content roughly correlates with Ca and is highest in the olivine/basanitic nephelinite macrocrysts (up to 0.007 apfu; Fig. 11b). The composition of some of these grains is typical of mantle olivines (e.g. Foley *et al.*, 2013; Bussweiler *et al.*, 2015). Olivine from carbonatite samples is distinctly higher in Mn (0.04–0.06 apfu) compared to olivine from the silicate rocks (≤ 0.01 apfu; Fig. 11c). Monticellite only

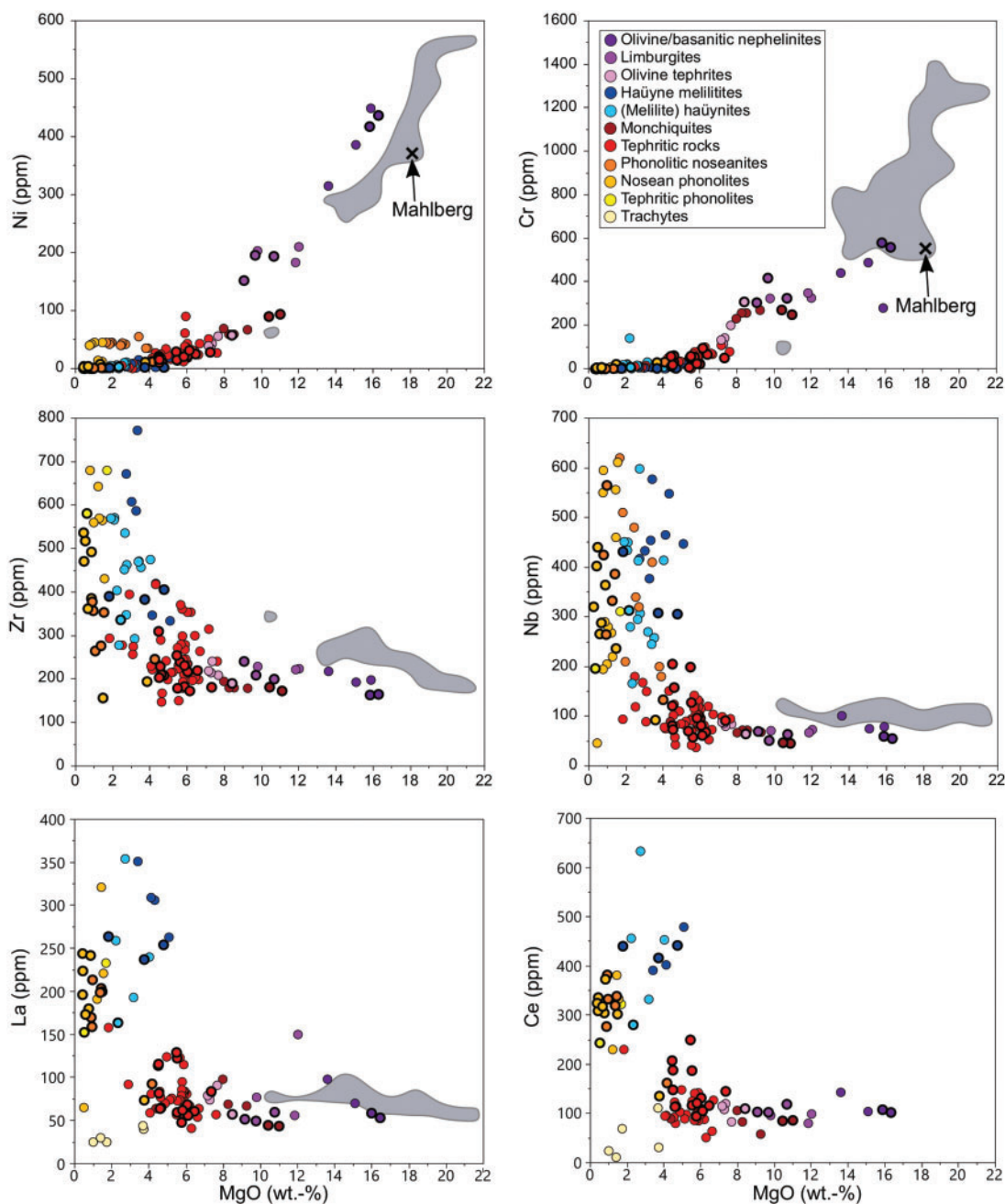


Fig. 9. Whole-rock trace element concentrations (in ppm) of Ni, Cr, Zr, Nb, La and Ce vs MgO (wt %). Olivine melilitites from SW Germany are shown as a grey field (data from Keller *et al.*, 1990; Wilson *et al.*, 1995).

occurs in one carbonatite sample (HTAC1356), with CaMgSiO_4 ranging between 75 and 77 mol % and Mn content between 0.05 and 0.06 apfu.

Clinopyroxene

Diopside and hedenbergite (\approx QUAD), aegirine and Tschermak's pyroxenes ($\text{CaFe}^{3+}\text{Fe}^{3+}\text{SiO}_6$, $\text{CaTiAl}_2\text{O}_6$ and CaAlAlSiO_6) are the most important clinopyroxene endmember components in the KVC rocks (Fig. 12a). All samples contain at least some pyroxenes with a

diopside component that exceeds 50 mol %, but especially in häüynolites, melteigites and the phonolitic rocks, clinopyroxene compositions evolve towards higher hedenbergite and aegirine contents (Fig. 12b). Except for late-stage clinopyroxenes, especially of the häüyne melilitites and nosean phonolites, pyroxenes of the tephritic phonolites have the lowest diopside component. Olivine/basanitic nephelinites and limburgites show the highest Ti concentrations, followed by the tephritic rocks. Those from nosean phonolites and tephritic phonolites have the lowest Ti values

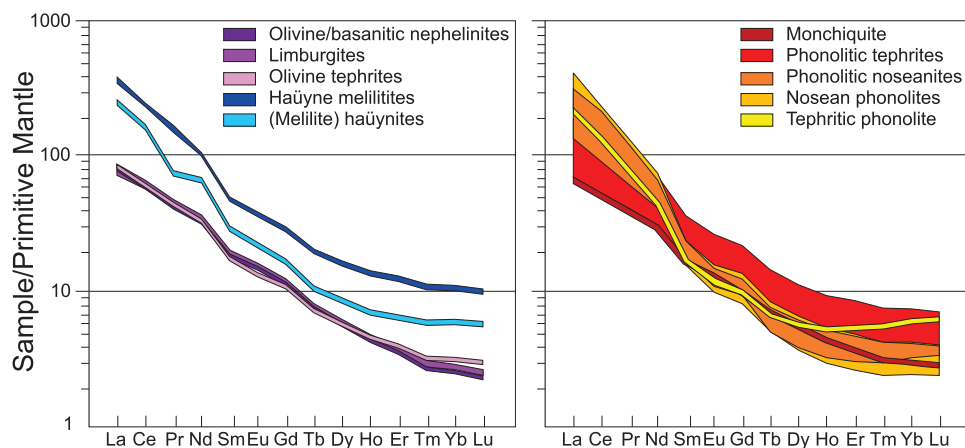


Fig. 10. Primitive mantle normalized rare earth element patterns for whole-rock analyses of various KVC rocks (normalizing values from [Palme & O'Neill, 2003](#)).

([Supplementary Data Table S3](#)). In some clinopyroxenes of the mafic rock groups, Cr is elevated, with the highest values (up to 0.04 apfu) observed in clinopyroxene from olivine/basanitic nephelinites.

Spinel group. Members of the spinel group in the KVC rocks vary in composition between rock types ([Fig. 13](#)). Limburgites, olivine tephrites ([Fig. 13a](#)) and monchiquites ([Fig. 13b](#)) contain Cr-rich spinel with $X_{Cr} [=Cr/(Cr+Al)]$ of up to 0.64 and Mg contents between 0.22 and 0.60 apfu ([Supplementary Data Table S4](#)). They mainly occur as inclusions in olivine, but are also present within the groundmass, where they either show a continuous or an abrupt (in monchiquites) decrease of the Cr- and a corresponding increase of the Ti-concentration from core to rim. These Ti-rich magnetites have higher ulvöspinel and spinel components ($Mag_{33-62}Usp_{25-57}Sp_{4-23}$) compared to the magnetites of phonolitic rocks, häüynolites and plagioclase-absent melteigites ($Mag_{7-91}Usp_{8-29}Sp_{0-3}$). Magnetites of the plagioclase-bearing melteigite ($Mag_{41-68}Usp_{27-51}Sp_{5-10}$) resemble those of the tephritic rocks ($Mag_{36-67}Usp_{25-57}Sp_{4-19}$). Magnetites from carbonatites have Mn contents ([Fig. 13a](#)) that are distinctively higher (up to 0.22 apfu) than those of the silicate rocks (up to 0.07 apfu). The content of Mg varies between 0 and 0.5 apfu and is lowest in the phonolitic rocks, whereas in the carbonatites, it locally reaches levels found in magnesioferrite.

Olivine/basanitic nephelinites and one sodalite monzogabbro contain primary ilmenite as intergrowths with magnetite, whereas ilmenite in the carbonatite sample HTAC1282 is present as individual grains. The composition varies between $Ilm_{57-76}Hem_{4-14}Pph_{1-20}Gk_{1-32}$ with the highest pyrophanite and lowest geikielite contents observed in the carbonatite.

Garnet

Garnets in the KVC rocks are andradite-schorlomite solid solutions ([Fig. 14a](#)), and the positive correlation between $TiFe^{3+}$ and $SiAl$ ([Fig. 14b](#)) illustrates the main substitution

process involving grossular (0–30%), andradite (50–90%) and schorlomite (0–38%) endmember components (calculated after [Locock, 2008](#)). Morimotoite (0–34%) represents the remaining component ([Fig. 14c](#)). The content of Zr (generally <0.03 apfu) is elevated in some garnets of the tephritic phonolites (up to 0.08 apfu; [Supplementary Data Table S5](#)), whereas the total amount of REE is always below 0.03 apfu. There is a general decrease of Mg and Ti from the core towards the rim, and based on these two elements, several compositional subgroups representing the different garnet-bearing lithologies can be distinguished ([Fig. 14d](#)).

Amphibole

Amphibole in sodalite monzogabbros and melanocratic phonolites is kaersutite, whereas Ti-poor amphibole in a phonolitic tephrite is magnesiopargasite ([Leake et al., 1997, 2004; Ridolfi et al., 2018 Fig. 15a](#)). Charge balance is maintained by coupled substitutions involving Na, K, Ti, Ca and Al ([Fig. 15b](#)), but is more complex in magnesiopargasites (HTAC1117), probably due to higher amounts of calculated Fe^{3+} (0.90–1.07 apfu versus 0.56–0.88 apfu; [Supplementary Data Table S6](#)). Amongst the halogens, F (up to 0.3 apfu) is distinctively higher than Cl (<0.04 apfu), reaching the highest values in sodalite monzogabbros. There is, however, no correlation between F and X_{Mg} [$Mg/(Mg+Fe+Mn+Ti+Al^{VI})$]; [Fig. 15c](#)].

Mica

Micas of the KVC form solid solutions consisting of phlogopite, annite, eastonite and siderophyllite ([Fig. 15d](#)). The Si+Al deficit in many analyses (except for the phonolitic tephrites) indicates incorporation of Ti or Fe^{3+} in the tetrahedral site ([Farmer & Boettcher, 1981](#)). Variable Ba contents (up to about 0.5 apfu; [Supplementary Data Table S7](#)) are incorporated by the coupled substitution $Ba + Al = Si + K$ ([Fig. 15e](#)). Similar to amphiboles (see above), Cl is <0.04 apfu while F reaches up to 1 apfu ([Fig. 15e](#)). The correlation between F and X_{Mg} [$Mg/(Mg+Fe+Mn+Ti+Al^{VI})$] for most of the analyses ([Fig. 1f](#))

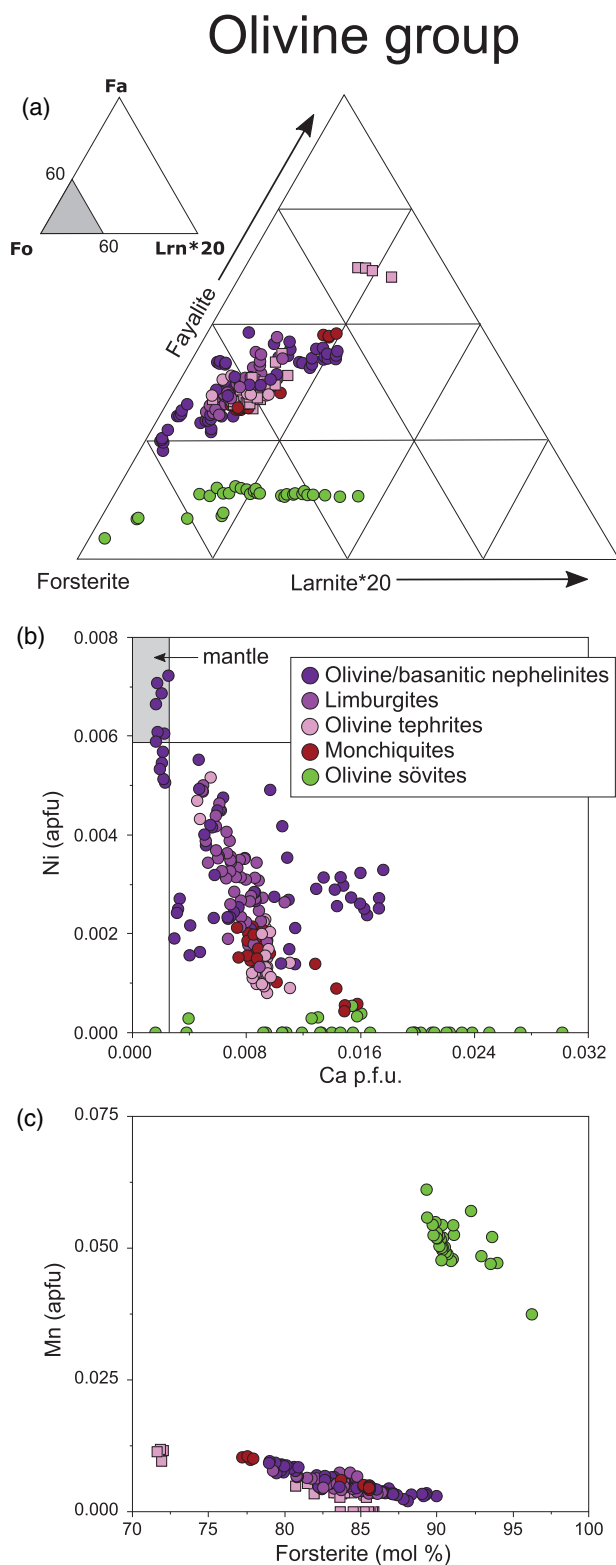


Fig. 11. Olivine compositions from the KVC rocks. (a) Forsterite–fayalite–larnite triangle with the larnite component being exaggerated by a factor of 20. (b) Contents of Ni (apfu) vs Ca (apfu). The upper left corner (grey) represents typical mantle olivine compositions (e.g. [Foley et al., 2013](#); [Busseweiler et al., 2015](#)). Rectangles are data from [Kim \(1985\)](#). (c) Amount of Mn (apfu) vs forsterite content (mol %).

implies the well-known F–Fe avoidance (e.g. [Munoz, 1984](#); [Speer, 1984](#); [Léger et al., 1996](#)). Exceptions correspond to very early (xeno-)crystic (phonolitic tephrite HTAC1117) and late-stage mica (häüyne melilitites).

Titanite and perovskite. Titanite shows a 2:1 correlation between Ti and (Al + OH + F) ([Fig. 16a](#)), indicating coupled substitutions $\text{Al} + \text{OH} = \text{Ti} + \text{O}$ and $\text{Al} + \text{F} = \text{Ti} + \text{O}$ (e.g. [Carswell et al., 1996](#); [Markl & Piazzolo, 1999](#); [Tropper & Manning, 2008](#)). However, Fe^{3+} is also relevant in this substitution process because the sum of Al and Fe^{3+} clearly correlates with F (up to 0.2 apfu in titanites from sodalite monzogabbro HTAC1343; [Fig. 16b](#); [Supplementary Data Table S8](#)). Titanite in the tephritic phonolites has the highest REE (up to 0.02 apfu), Nb (up to 0.07 apfu) and Zr (up to 0.07 apfu) contents. Late-stage titanite, which formed at the expense of garnet (wollastonite-bearing nosean phonolite HTAC1345, see above) incorporates the highest amounts of Sr (up to 0.026 apfu). In perovskites, minor substitutions of Ca with Na (up to 0.044 apfu; [Supplementary Data Table S9](#)) and REE (up to 0.029 apfu), and Ti with Al (up to 0.007 apfu) and Nb (up to 0.074 apfu) were detected, whereas Zr is generally below the detection limit.

Feldspar

Plagioclase is generally unzoned and varies in composition from $\text{An}_{82}\text{Ab}_{17}\text{Or}_1$ to $\text{An}_{22}\text{Ab}_{70}\text{Or}_8$. The highest anorthite component is present in some phonolitic tephrites and the lowest in tephritic phonolites ([Fig. 17a](#)). Olivine tephrites and tephritic phonolites contain plagioclase with the highest Sr contents (up to 0.056 apfu; [Supplementary Data Table S10](#)), whereas the remaining feldspar analyses (including alkali feldspar) show a broad range down to the detection limit. Alkali feldspar composition ranges from $\text{An}_{10}\text{Ab}_{65}\text{Or}_{25}$ to $\text{An}_0\text{Ab}_{03}\text{Or}_{97}$. In alkali feldspar from phonolitic rocks, Ba may reach up to 0.18 apfu; in the remaining rock types, Ba is generally <0.04 apfu.

Nepheline

Nepheline composition varies between $\text{Ne}_{70}\text{Ks}_{27}\text{Qtz}_{03}$ and $\text{Ne}_{82}\text{Ks}_{17}\text{Qtz}_{01}$. The Ca content is lower (0.02–0.13 apfu; [Supplementary Data Table S11](#)) in häüyne melilitites, melilite häüynites and phonolitic noseanites than in the melteigites (0.13–0.25 apfu). The highest Fe^{3+} concentrations (0.16 apfu) occur in nepheline from häüyne melilitites and melilite häüynites, while the lowest (0.01 apfu) were observed in the melteigites.

Sodalite group. Minerals of the sodalite group ([Fig. 17b](#)) are Cl-dominated (sodalite-rich) in tephritic rocks, but mostly S-rich (nosean and häüyne) in phonolitic and in melilitic to häüynitic rocks, respectively ([Fig. 18](#)). In melilitic to häüynitic rocks, large grains have distinctly higher S contents than smaller grains and show a

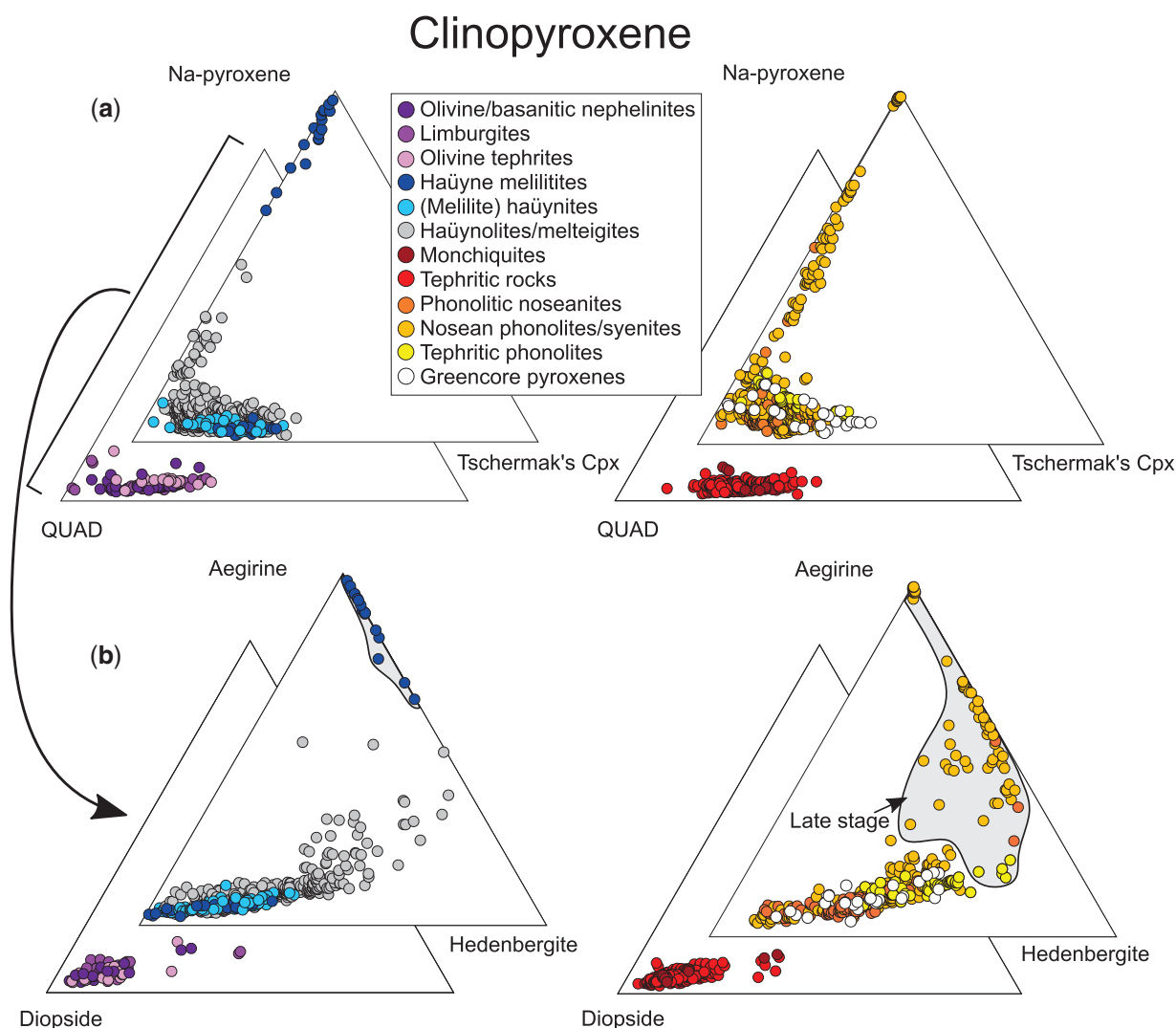


Fig. 12. Pyroxene compositions from the KVC rocks. (a) Na-pyroxene ($\text{NaFeSi}_2\text{O}_6+\text{NaAlSi}_2\text{O}_6$)-QUAD ($\text{CaMgSi}_2\text{O}_6+\text{CaFeSi}_2\text{O}_6$)-Tschermak's pyroxene ($\text{CaFeFeSiO}_6+\text{CaTiAl}_2\text{O}_6+\text{CaAlAlSiO}_6$) triangles. (b) Diopside-hedenbergite-aegirine triangles.

decrease in S from core to rim. The contents of K and Fe reach up to 1.0 and 0.4 apfu, respectively (Supplementary Data Table S12).

Melilite group. Melilite in häüyne melilitites has a higher akermanite and a lower soda melilite component compared to melilite häüynites (Fig. 17c). In häüyne melilitites, melilite phenocrysts have slightly higher MgO and CaO concentrations (Supplementary Data Table S13) than the groundmass grains.

DISCUSSION

Crystallization conditions of the KVC rocks

In this section, various geothermobarometers are applied to the KVC rocks to derive estimates of the crystallization conditions (P , T , $f\text{O}_2$, $a\text{SiO}_2$) of the various lithologies. These are then combined with fractional

crystallization calculations to reconstruct the petrological evolution of this composite volcano.

Depth of emplacement

The only means to estimate the emplacement depth of the KVC rocks is via Al-in amphibole barometry, requiring equilibrium amphibole-plagioclase assemblages and independent T estimates (Anderson & Smith, 1995). The results of these calculations depend on redox-sensitive Tschermak and Fe^{3+} -Al substitutions in the amphibole structure, and it has been suggested to apply this barometer only to specific amphibole compositions with intermediate Fe contents and significant calculated Fe^{3+} ($\text{Fe}_{\text{tot}}/[\text{Fe}_{\text{tot}}+\text{Mg}] = 0.40\text{--}0.65$; $[\text{Fe}^{3+}/[\text{Fe}^{3+}+\text{Fe}^{2+}] > 0.25$), otherwise pressure may be overestimated (Anderson & Smith, 1995). Although some KVC amphibole compositions do not fulfil all requirements, we applied this barometer since the investigated

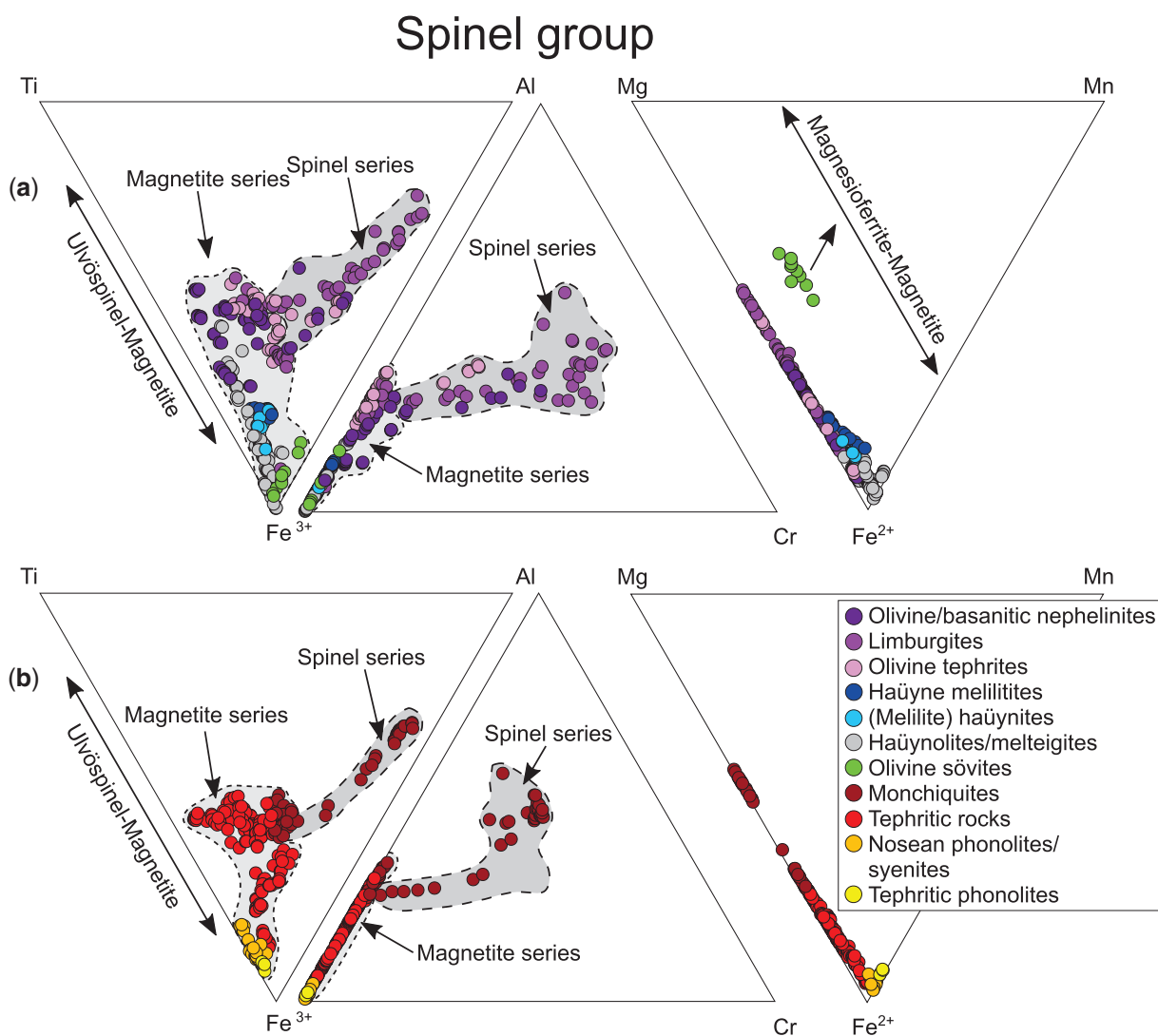


Fig. 13. Spinel compositions from the KVC rocks. (a) Nephelinitic to limburgitic rocks, melilititic to häüynitic rocks and olivine sövites. (b) Tephritic to phonolitic rocks.

amphiboles formed at high redox conditions (see below).

Four suitable samples of phonolitic tephrite and sodalite monzogabbro yielded a temperature of 920°C (two feldspar solvus thermometry; Fuhrman & Lindsley, 1988) and a calculated pressure range between 0.1 to 1.7 kbar (uncertainty of ± 0.6 kbar). The highest values derive from a sample that contains abundant 'green-core' clinopyroxenes (HTAC1117), implying that some clinopyroxenes (and amphiboles) from this sample may have been entrapped at greater depth (or may have formed at higher temperature). For the remaining samples, pressures <0.9 kbar (average of 0.3 kbar) were calculated. In these samples, pressures calculated from amphibole rim analyses are lower than those calculated from core analyses (Fig. 18a). The calculated pressure range may reflect continuous crystallization during ascent from a magma chamber over a distance of about 2–4 km to sub-surface emplacement. Thus, a shallow level plumbing system not very

different from what is assumed for Mt. Vulture (e.g. Beccaluva *et al.*, 2002) can be envisaged, with potential magma storage at shallow sub-surface levels. Temperature calculation according to the thermometer of Holland & Blundy (1994) strongly depends on pressure, but overlaps with the results of the two-feldspar thermometer (Fuhrman & Lindsley, 1988) if a mean pressure of 0.3 kbar and no Ti on the tetrahedral site of the amphibole is assumed. This pressure was used for all subsequent calculations.

Estimates of T - $a\text{SiO}_2$ - $f\text{O}_2$ conditions

Olivine–clinopyroxene–magnetite assemblages in olivine/basanitic nephelinites and the slightly more evolved limburgites and olivine tephrites yield equilibration temperatures of 1000–1050°C, 990–1030°C, and 1000–1150°C, respectively (Fig. 18b and Supplementary Data Table S14). We interpret these as near-liquidus temperatures. For the same rocks, the clinopyroxene–liquid

Garnet

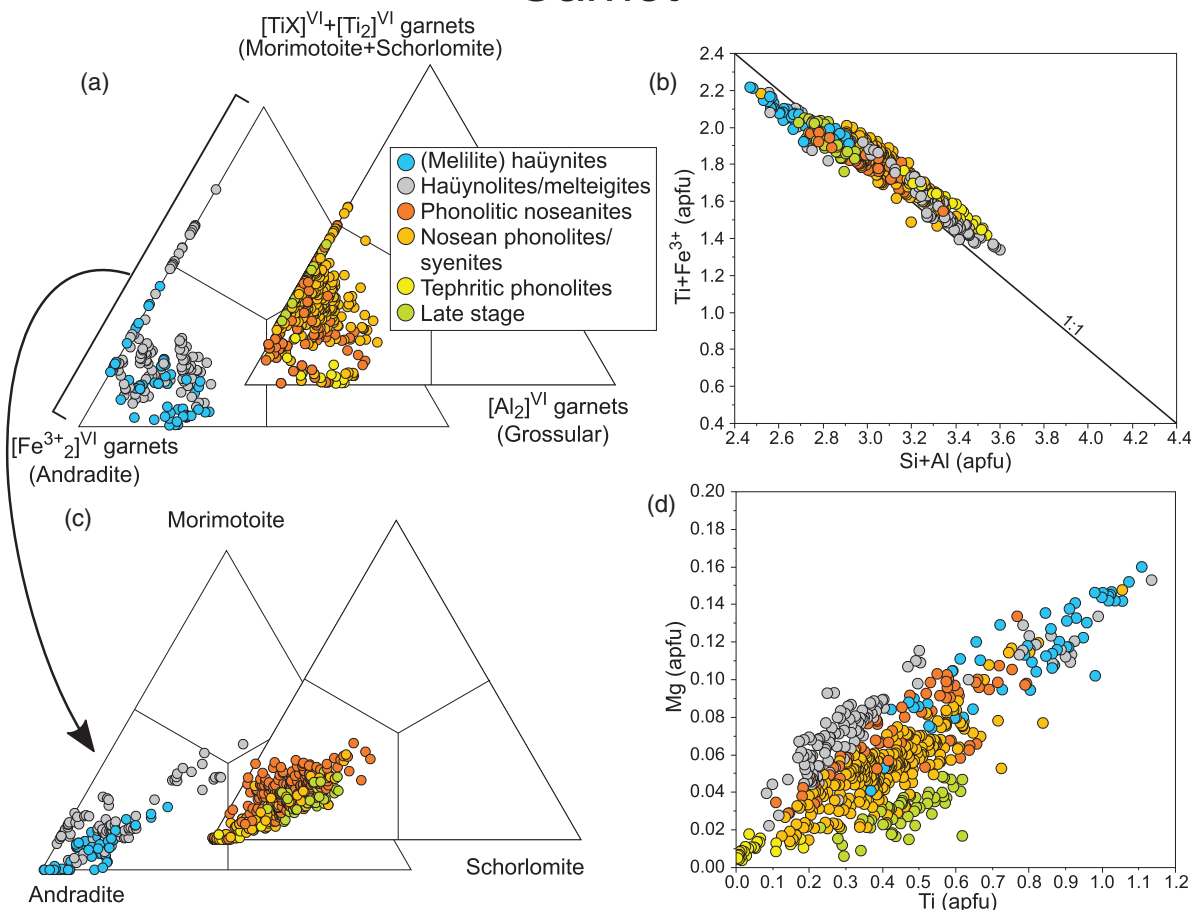


Fig. 14. Correlation and compositions diagrams for garnet from the KVC rocks. (a) Garnet composition based on the octahedral-site occupancy. (b) Ti+Fe^{3+} apfu vs Si+Al apfu diagram illustrating a good 1:1-correlation (solid line). (c) Garnet composition expressed as andradite-morimotoite-schorlomite endmembers. (d) Mg apfu vs Ti apfu subdivides various garnet types and lithologies into specific groups.

thermometer of [Putirka *et al.* \(1996\)](#) indicates higher temperatures (1130–1260°C). However, these values have to be treated with caution, because the whole-rock data were used as proxies for melt compositions and in some samples, xenocrysts of olivine and clinopyroxene were identified based on characteristic textures ([Uljanov *et al.*, 2007](#); see above). Calculated $a\text{SiO}_2$ values for this rock group ([Fig. 18c](#)) increase from olivine nephelinites ($a\text{SiO}_2=0.4\text{--}0.5$), through the basanitic nephelinite ($a\text{SiO}_2=0.7\text{--}0.8$) to limburgites and olivine tephrites (0.5–0.9). This is in agreement with the absence of feldspar in the former but its appearance in the latter and can be explained by fractionation of olivine (see below). Redox conditions are estimated to around $\Delta\text{FMQ}^* = +2\text{--}3$ ([Fig. 18c](#)). However, application of the magnetite-melt oxybarometer of [Arató & Audétat \(2017\)](#) reveals slightly lower values ($\Delta\text{FMQ}^* = +1\text{--}2$), which may be related to the methodological limitations mentioned above.

For häüyne melilitites and slightly more evolved (melilite) häüynites similarly high crystallization

temperatures were estimated (1060–1120°C and 950–1010°C, respectively; [Fig. 18b](#) and [Supplementary Data Table S14](#)). Their estimated silica activity, however, is very low ($a\text{SiO}_2 < 0.22$) and highly oxidized conditions ($\Delta\text{FMQ}^* = +4\text{--}6$) are determined ([Fig. 18c](#)). This is in accordance with their mineralogy, i.e. the abundance of melilite, feldspathoid minerals and garnet, whilst olivine and feldspar are scarce or absent. Due to the absence of suitable phases, no constraints on the crystallization temperature of coarse-grained foidolitic cumulates could be calculated. The absence of K-rich minerals, the occurrence of perovskite and the composition of the sodalite-group minerals in the foidolitic xenoliths suggest that these rocks are rather related to the häüyne melilitites and häüynites than to the phonolitic rock group. Based on their mineralogical similarity, a comparable crystallization history as for häüyne melilitites and (melilite) häüynites can be assumed. The minimum silica activity of titanite-bearing and perovskite-lacking häüynolites is around 0.1, the transition from perovskite to titanite in some melteigites

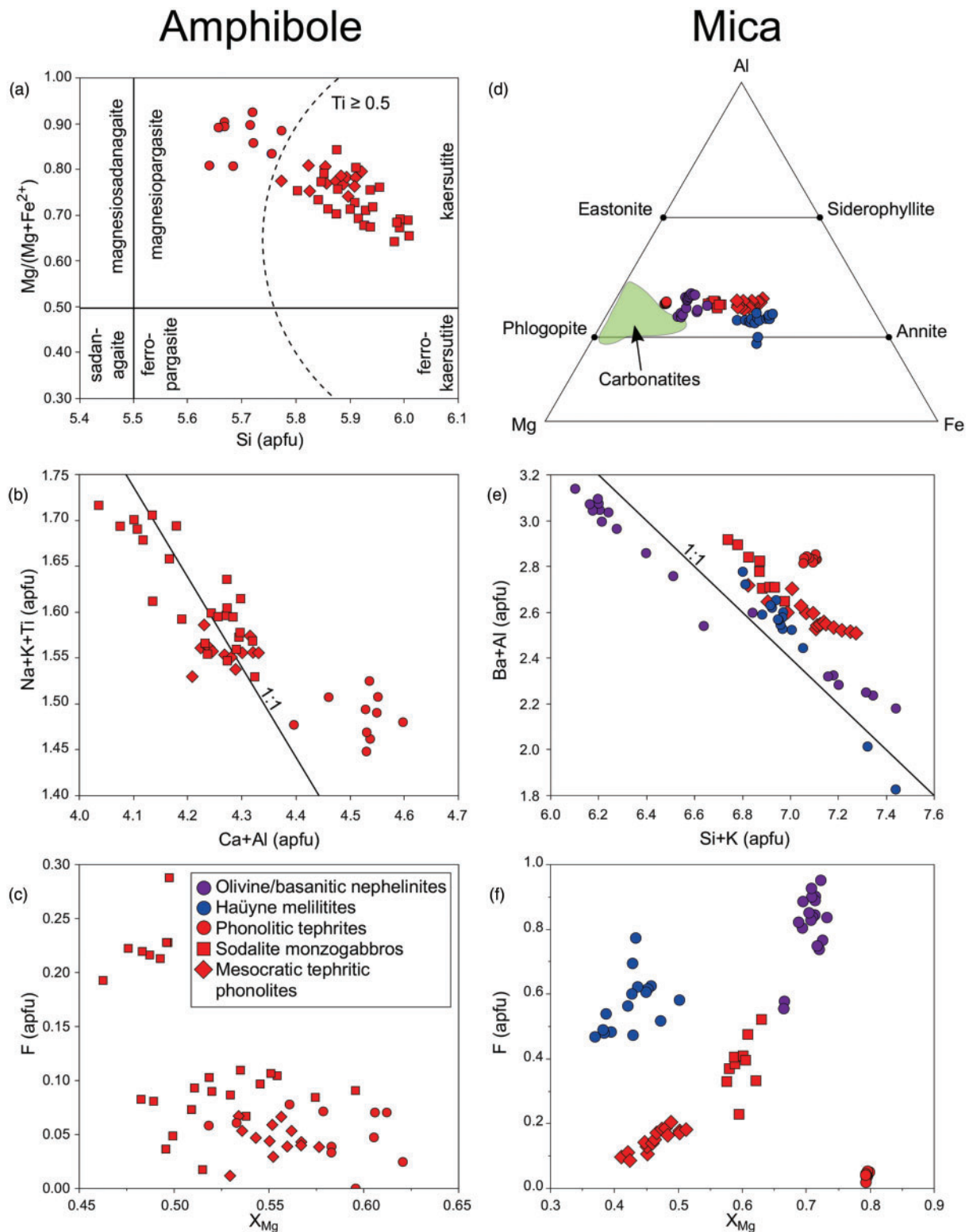


Fig. 15. (a) Classification and correlation diagrams for amphibole (a–c) and mica (d–f) from the KVC rocks. (a) Amphiboles are classified as magnesiopargasite or kaersutite (after Leake *et al.*, 1997, 2004). (b) Correlation of $\text{Na}+\text{K}+\text{Ti}$ apfu vs $\text{Ca}+\text{Ba}+\text{Al}$ apfu (c) Correlation diagram of F apfu vs X_{Mg} ($\text{Mg}/(\text{Mg}+\text{Fe}+\text{Mn}+\text{Ti}+\text{Al}^{\text{VI}})$). (d) Mica composition based on the amount of Mg , Al and Fe . Green area represents the compositional range of carbonatites at the KVC (unpublished data). (e) Diagram showing a good 1:1 correlation between $\text{Ba}+\text{Al}$ apfu and $\text{Si}+\text{K}$ apfu (f) Correlation diagram of F apfu vs X_{Mg} ($\text{Mg}/(\text{Mg}+\text{Fe}+\text{Mn}+\text{Ti}+\text{Al}^{\text{VI}})$).

Titanite

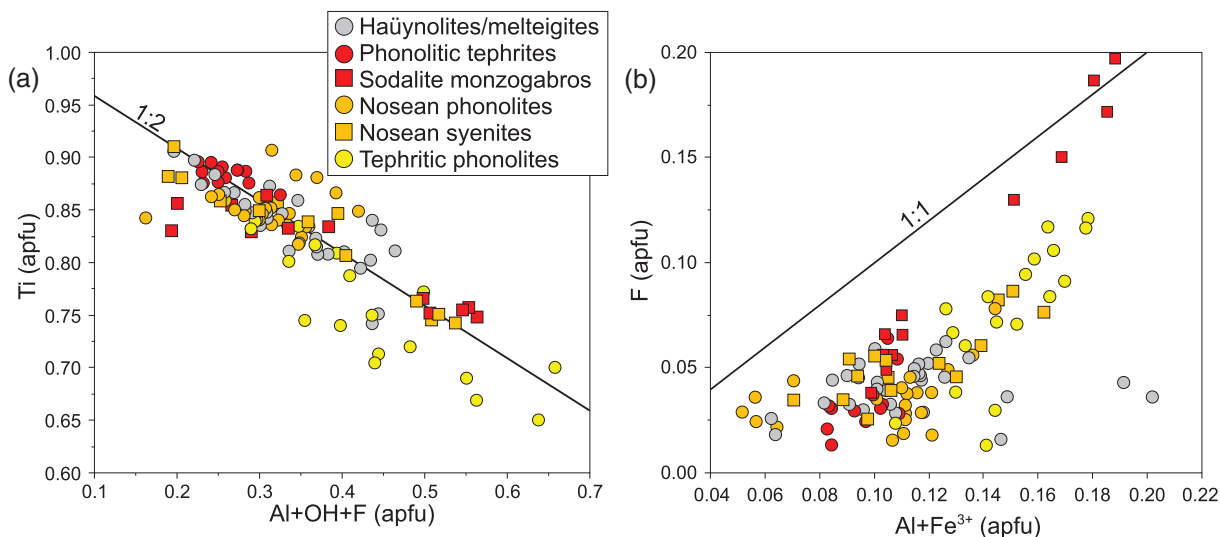


Fig. 16. Titanite compositions from the KVC rocks. (a) Correlation of Ti apfu vs Al+OH+F apfu (b) Correlation of Al+Fe³⁺ apfu vs F apfu, illustrating the additional influence of Fe³⁺.

(Fig. 3d) suggests $a\text{SiO}_2$ values of 0.1–0.15. High $f\text{O}_2$ conditions ($\Delta\text{FMQ}^* = +3$ –6; Fig. 18c), similar to häüyne melilitites and (melilite) häüynites are indicated.

Tephritic to phonolitic rocks reveal variable T– $a\text{SiO}_2$ – $f\text{O}_2$ conditions. Estimated near-liquidus temperatures decrease with evolution from primitive monchiquites (~1040–1100°C) towards tephrites and phonolites (~960–880°C), based on QUILF calculations, two-feldspar thermometry (Fig. 17a) and hornblende-plagioclase thermometry (Supplementary Data Table S14). Again, the clinopyroxene-liquid thermometer of Putirka *et al.* (1996) generally yields distinctively higher temperatures (up to 1220°C), as do melt inclusion-homogenization data (up to 1170°C; Panina *et al.*, 2000). Silica activity decreases (Fig. 18c) from monchiquites ($a\text{SiO}_2 = 0.6$ –0.8) via tephritic ($a\text{SiO}_2 = \sim 0.5$) to phonolitic rocks ($a\text{SiO}_2 = \sim 0.2$), probably as a consequence of intense clinopyroxene fractionation (see below). Estimated $f\text{O}_2$ (Fig. 18c) increases from monchiquites ($\Delta\text{FMQ}^* = +2$ –3) towards tephritic ($\Delta\text{FMQ}^* = +3$ –4) and phonolitic rocks ($\Delta\text{FMQ}^* = +3$ –5). This is in accordance with the change of the mafic liquidus assemblage from olivine–clinopyroxene–spinel in primitive rocks (monchiquites) to clinopyroxene–garnet–titanite in the more evolved rocks (compare Marks *et al.*, 2008a). Similarly, the shift from sulfate-poor sodalite-group minerals in tephrites towards sulfate-rich members (nosean/häüyne) in the more evolved phonolitic rocks (Fig. 17b), paralleled by a decrease in ulvöspinel component in magnetites (Fig. 13), implies increasingly oxidized conditions.

For carbonatites, the crystallization temperature is constrained to 840–900°C (Fig. 18c and Supplementary Data Table S14) based on the Fe–Mg exchange reaction between olivine and magnetite (QUILF). Reaction (6) allows determination of a maximum $a\text{SiO}_2$ value of

0.05, using a monticellite endmember component of 0.8 and a diopside activity of 1. However, monticellite is a very rare phase in the KVC carbonatites and diopside is absent in the investigated samples. Therefore, this estimate may not be representative of all carbonatites in the KVC. Nevertheless, $a\text{SiO}_2$ in carbonatites is certainly low, and perovskite-bearing carbonatites record an upper silica activity limit of 0.2. Redox estimates for the investigated samples ($\Delta\text{FMQ}^* = +6$ –7) are distinctively higher than previous $f\text{O}_2$ estimates on other carbonatites, which fall between $\Delta\text{FMQ} = -0.8$ and +1.0 (Friel & Ulmer, 1974; Gaspar & Wyllie, 1983; Treiman & Essene, 1984; Ryabchikov *et al.*, 2008; Milani *et al.*, 2016). This apparent discrepancy is caused by the fact that the cited publications present ΔFMQ calculations at $a\text{SiO}_2$ values of 1. However, if they are corrected to low silica activities (=FMQ*, as is done here), much more oxidized values emerge. The association of carbonatites with quite oxidized silicate rocks is not uncommon at other localities, such as Mt. Vulture, Italy (e.g. De Fino *et al.*, 1982; Beccaluva *et al.*, 2002; Panina & Stoppa, 2009), Magnet Cove, USA (e.g. Flohr & Ross, 1990), Tamazeght, Morocco (e.g. Marks *et al.*, 2008a), Oldoinyo Lengai (e.g. Zaitsev *et al.*, 2012) or Kerimasi (e.g. Guzmics *et al.*, 2012). In case of the KVC, carbonatites represent the most oxidized lithologies, but it remains to be tested if this is true for other carbonatite complexes.

Genetic relationships between the various KVC rocks

Based on their MgO, Ni and Cr contents (Figs 8 and 9), olivine/basanitic nephelinites, limburgites and monchiquites represent the most primitive rock types of the KVC. Their contrasting mineralogy, mineral chemistry, petrology, and geochemistry (see above),

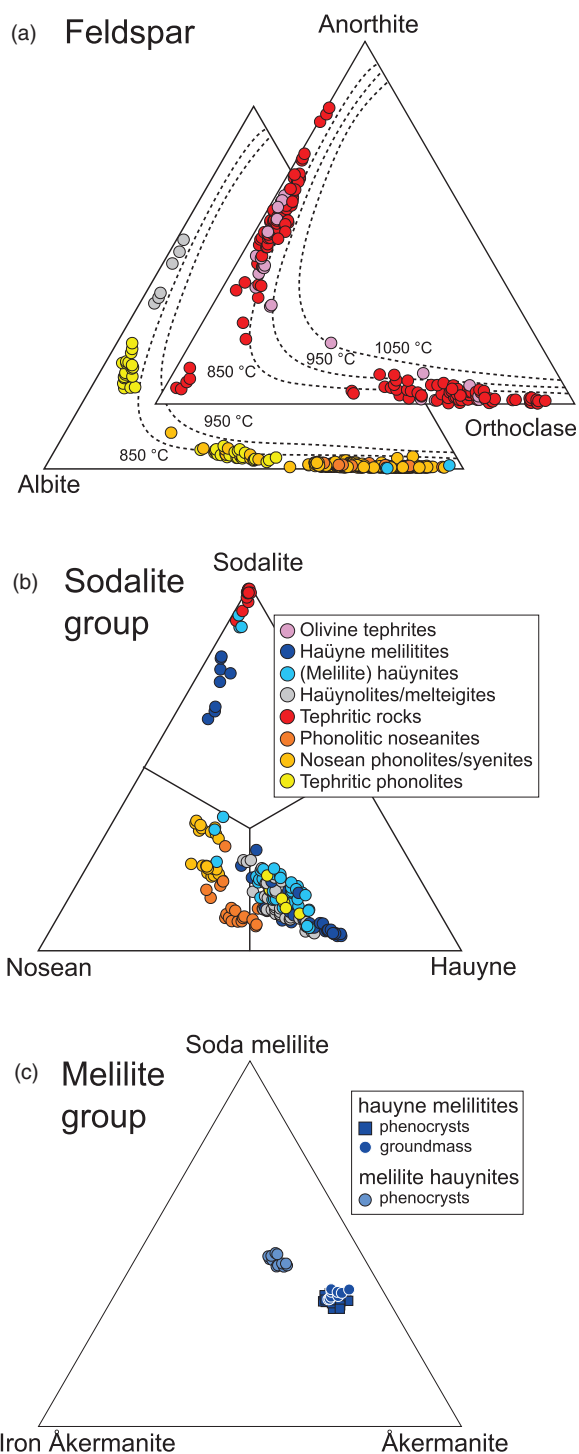


Fig. 17. (a) Feldspar compositions in the KVC rocks illustrated in the albite-anorthite-orthoclase triangle. Isotherms (dashed lines) represent the feldspar solvus after Fuhrman & Lindsley (1988). (b) Nosean-sodalite-haüyne triangle showing the compositional range of sodalite-group minerals. (c) Melilite composition of häüyne melilitites and melilite häüynites in the iron åkermanite-soda melilite-åkermanite triangle.

however, makes a common parental magma unlikely. Monchiquites and olivine/basanitic nephelinites probably represent the most primitive members of at least two distinct parental magmas, as previously suggested

by Keller (1984b) and Schleicher *et al.* (1990). The association of feldspar-free nephelinitic/melilititic rocks with plagioclase-normative, primitive basanitic/tephritic rocks is a common association world-wide (e.g. Melluso *et al.*, 2016) and is, for example, very similar to the rock associations known from Mt. Vulture (Italy), including the occurrence of associated calciocarbonatite (e.g. De Fino *et al.*, 1982; Melluso *et al.*, 1996; Rosatelli *et al.*, 2000, 2007; Beccaluva *et al.*, 2002; Stoppa *et al.*, 2008). We tested the importance of fractional crystallization of monchiquitic/basanitic and nephelinitic/melilititic magmas for producing the lithological and geochemical variation observed in the KVC rocks by means of mass balance calculations using phenocryst assemblages as potential fractionating phases (Supplementary Data Table S15 and Fig. 2). The results of these calculations are discussed in the following sections.

A model for the evolution of tephritic to phonolitic rocks

The evolution from tephritic to phonolitic rocks is mirrored by an increasing hedenbergite component in clinopyroxene (Fig. 12), decreasing Ti and Mg contents in magnetite (Fig. 13) and decreasing Mg concentrations in garnet (Fig. 14), and is further strengthened by whole-rock data (Figs 7 and 8). The evolution from monchiquites towards tephrites can be modelled by the fractionation of clinopyroxene and minor amounts of olivine and spinel. Further fractionation of clinopyroxene and magnetite explains the evolution towards phonolitic tephrites and the compositional variation of most phonolites requires further fractionation of clinopyroxene, plagioclase, häüyne/nosean, as well as garnet and titanite. The importance of garnet and titanite fractionation at this stage is further supported by whole-rock REE systematics. Except for an increasing total amount of REE, the REE pattern does not change during evolution from monchiquites via tephrites to phonolitic tephrites (Fig. 10), as the fractionating phases incorporate negligible amounts of REE (Schnetzler & Philpotts, 1970; Grutzeck *et al.*, 1974; Hanson, 1980). With the onset of garnet and titanite crystallization, this situation changes; as garnet prefers the heavy REE (e.g. Schnetzler & Philpotts, 1970; Shimizu & Kushiro, 1975; Nicholls & Harris, 1980; Marks *et al.*, 2008b), garnet fractionation leads to a depletion of heavy REE in the remaining melt. Titanite has high Kd values for REE and preferentially incorporates the middle REE (e.g. Simmons & Hedge, 1978; Prowatke & Klemme, 2005). Accordingly, total REE concentration decreases and the resulting REE patterns for evolved tephritic phonolites are trough-shaped (Figs 10 and 19). Similarly, Zr/Hf ratios are almost constant in monchiquites and phonolitic tephrites, but strongly increase towards the phonolitic rocks (Fig. 19). The Nb/Ta and Y/Ho ratios behave in a similar way, except for four phonolitic tephrites, which show anomalous high ratios. (Note that these samples

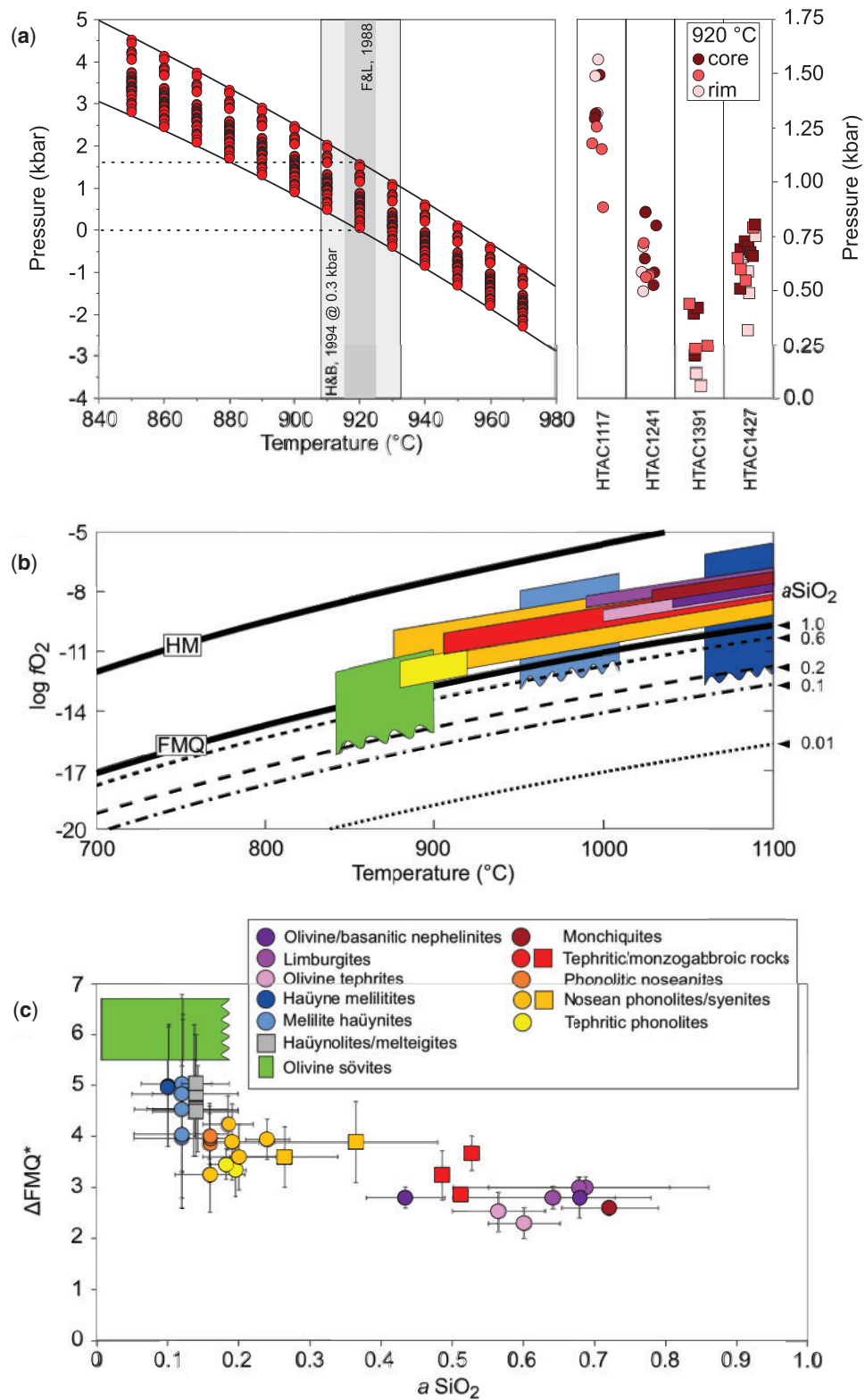


Fig. 18. Estimated crystallization conditions for the KVC rocks. (a) Pressure-temperature dependence (left side) of the Al-in hornblende barometer of [Anderson & Smith \(1995\)](#). Dark grey area represents the temperature interval calculated using the feldspar liquidus thermometer of [Fuhrman & Lindsley \(1988\)](#), bright grey area shows the temperature range obtained using the amphibole-plagioclase thermometer of [Holland & Blundy \(1994\)](#) at 0.3 kbar. On the right side, pressure estimates at 920°C are shown. Plutonic rocks are shown as rectangles whereas volcanic rocks are displayed as circles. (b) Temperature vs log fO_2 diagram with the FMQ and HM buffers curves shown as thick lines, whereas dashed lines illustrate the FMQ buffer curve at variable silica activities. (c) Estimated silica activity (a_{SiO_2}) vs oxygen fugacity (expressed as ΔFMQ^*) for the KVC. Symbols illustrate mean values whereas the error bars indicate the total possible range.

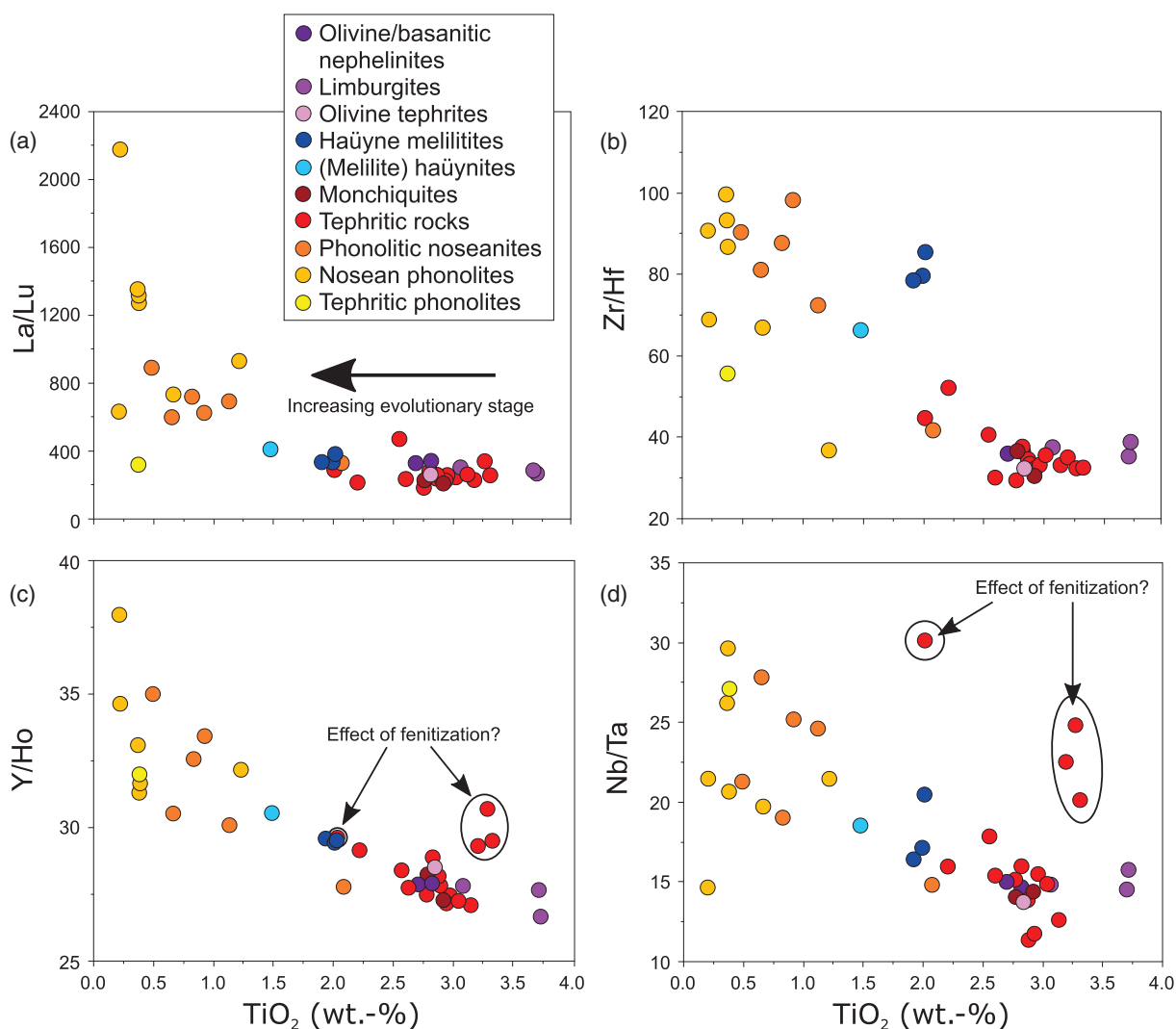


Fig. 19. Whole-rock La/Lu (a), Zr/Hf (b), Y/Ho (c) and Nb/Ta (d) vs TiO_2 illustrating the effects of titanite, perovskite and garnet fractionation (see text for details).

were collected in the immediate vicinity of the large sövite bodies and show subtle petrographic evidence for fenitization). Crystallization of titanite/perovskite and garnet are known to cause such fractionation effects (e.g. Chakhmouradian, 2006; Marks *et al.*, 2008b; Olin & Wolff, 2012; Chakhmouradian *et al.*, 2013).

Phonolitic rocks of the KVC with $\text{Na}_2\text{O}+\text{K}_2\text{O}$ concentrations > 12 wt % cannot be explained by fractionation of the observed mineral phases and their chemical composition (Supplementary Data Fig. S2). We suggest that these rocks accumulated significant nosean and alkali feldspar (Supplementary Data Table S15). For some of these rocks, Sutherland (1967) proposed the importance of a metasomatic overprint and, in fact, some of them show abundant hydrothermally-formed zeolites (e.g. Weisenberger *et al.*, 2014), which could be responsible for their high and variable $\text{Na}_2\text{O}+\text{K}_2\text{O}$ contents. The potential importance of contamination with 'crustal material' for some of these rocks has been stressed

previously (Schleicher *et al.*, 1990) and mixing of variably evolved magmas and uptake of already crystallized material during the genesis of this rock series seems likely, as suggested, e.g. by 'green-core' pyroxenes and rounded olivine (xeno)crysts in some of the rocks (see above; Ulianov *et al.*, 2007). All these factors add geochemical complexity and need to be invoked to fully explain the compositional scatter in the whole-rock data.

A model for the evolution of nephelinitic to limburgitic and melilitic to häüynitic rocks. Olivine/basanitic nephelinites are the most MgO- Ni- and Cr-rich rocks of the KVC (Figs 8 and 9) and olivine melilitites (grey field in Figs 8 and 9) occurring in close proximity to the KVC (e.g. Mahlberg, Urach and Hegau) are interpreted to be their parental magmas (Wilson *et al.*, 1995; Dunworth & Wilson, 1998; Ulianov *et al.*, 2007). Olivine/basanitic nephelinites can be modelled assuming fractionation of olivine, melilite and minor amounts of spinel,

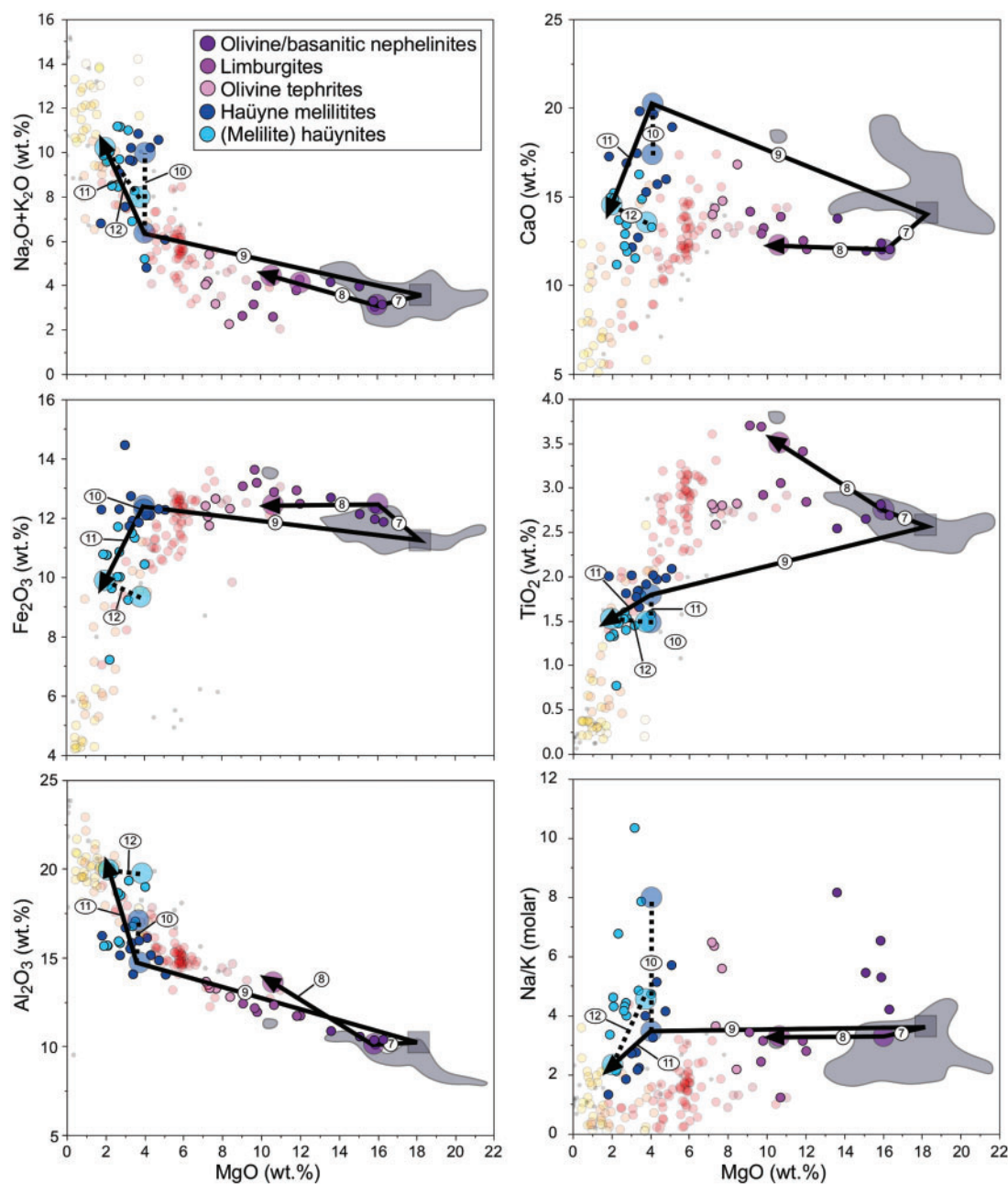


Fig. 20. Fractionation path for the nephelinitic to limburgitic and melilititic to haüynitic rock series. Numbers refer to [Supplementary Data Table S15](#). Grey area illustrates the compositional range of olivine melilitites in SW Germany ([Keller *et al.*, 1990](#); [Wilson *et al.*, 1995](#)).

perovskite and nepheline, using the Mahlberg olivine melilitite ([Fig. 1b](#)) as a starting composition; limburgites can be explained by further fractionation mainly of olivine and clinopyroxene ([Fig. 20](#) and [Supplementary Data Table S15](#)). However, their highly variable Na/K ratios ([Fig. 8](#)) cannot be easily reproduced by such fractionation models ([Fig. 20](#)), which we suggest is due to variable alteration and weathering of nepheline and glass, and accompanying redistribution of Na and K (e.g. [Nesbitt *et al.*, 1980](#); [Quantin & Lorenzoni, 1992](#); [Giampaolo *et al.*, 1997](#); [Weisenberger *et al.*, 2014](#)).

It has been suggested that, based on their (slightly) different Pb and Nd isotope compositions ([Schleicher *et al.*, 1990, 1991](#)), the olivine melilitites, olivine/basanitic nephelinites and limburgites were derived from a different mantle source compared with haüyne melilitites and (melilite) haüynites. However, (i) Pb concentrations in these rocks are very low (<35 ppm) and could be easily altered during the ascent of such melts, as proposed by [Ray \(2009\)](#) who assumed assimilation with lower crustal granulites for the KVC rocks; (ii) considering a typical uncertainty of about 0.5 ϵ_{Nd} units (e.g.

Marks *et al.*, 2003, 2009), the Nd isotope compositions of these two groups largely overlap with each other; (iii) the $^{87}\text{Sr}/^{86}\text{Sr}$ values for both rock groups are very similar (Schleicher *et al.*, 1990). Furthermore, an isotopically heterogeneous mantle could also be responsible for the isotope variations (e.g. Bell, 1998). We suggest that nephelinitic to limburgitic and melilititic to haüynitic rocks may be derived from a common source. In fact, haüyne melilitites could have originated from olivine melilitites by fractionation of greater quantities of olivine and lesser quantities of melilite (Fig. 20 and Supplementary Data Table S15) and further fractionation of mainly melilite from the haüyne melilitites could generate melt compositions similar to (melilite) haüynites. The large scatter in Na/K ratios, $\text{Na}_2\text{O}+\text{K}_2\text{O}$ and CaO contents in the haüyne melilitites and (melilite) haüynites (Figs 8 and 20) can be explained by the accumulation of variable amounts of haüyne, clinopyroxene and melilite, combined with the compositional variation and variably intense alteration of abundant feldspathoid minerals, similar to what we propose for the phonolitic rocks (see above).

High CO_2 concentrations in a silica-undersaturated magma lead to a polymerization of the silicate melt (Mysen *et al.*, 1976) and hence, more polymerized minerals like melilite compared to olivine should crystallize. Since the solubility of CO_2 depends on pressure, olivine/basanitic nephelinitic or haüyne melilititic melts may reflect polybaric melting of a common mantle source (Mysen *et al.*, 1976; Yagi & Onuma, 1978). However, in addition to pressure, the solubility of CO_2 in a melt may also be influenced by the oxidation state (e.g. Pan *et al.*, 1991; Thibault & Holloway, 1994). Highly oxidized conditions may cause the release of CO_2 and a subsequent termination of melilite precipitation (Mysen *et al.*, 1976). For example, Dunworth & Wilson (1998) suggested variable redox conditions to be responsible for the highly variable melilite-olivine ratio in olivine melilitites in southwestern Germany. Since haüyne melilitites crystallized under more oxidized conditions than olivine/basanitic nephelinites (Fig. 18c), it seems likely that haüyne melilitites originated from a more oxidized part of the mantle. However, during fractionation from olivine melilitite to haüyne melilitite, the CO_2 concentration must increase, as melilite and magmatic carbonate are present in the haüyne melilitites.

The genetic relation between silicate rocks and carbonatites

The similarity of carbon and oxygen isotope compositions of calcite crystals in sövites and haüyne melilitites imply a common origin (Hubberten *et al.*, 1988). Apatite textures and compositions in haüyne melilitites suggest that these may have nucleated in a silicate melt and continued to crystallize from a melt with carbonatitic affinity (Wang *et al.*, 2014). Further, similar $^{87}\text{Sr}/^{86}\text{Sr}$ isotope values of carbonatites and haüyne melilitites (Schleicher *et al.*, 1990) strengthen this hypothesis. As both the

carbonatites and the haüyne melilitites formed at high redox conditions (Fig. 18), their close genetic affinity is further supported, although the actual process of carbonatite formation is still uncertain: are they direct mantle carbonate melts or are they products of fractionation/liquid immiscibility of mantle-derived silicate melts?

A primary melt directly derived by (carbonated) mantle melting seems unlikely, as most of the KVC carbonatites are calcitic (Keller, 1978, 1981; Lehnert, 1989), whereas primary carbonatitic magmas should be Mg-rich (e.g. Wallace & Green, 1988; Dalton & Wood, 1993; Dalton & Presnall, 1998; Wyllie & Lee, 1998; Litasov & Ohtani, 2009). However, during ascent, any primary carbonatitic magma should evolve towards more calcitic compositions by interaction with mantle peridotite and accompanying wehrlitization (e.g. Dalton & Wood, 1993; Lee & Wyllie, 2000). In contrast, the observed $\text{Ca}/(\text{Ca}+\text{Mg})$ ratios of the KVC carbonatites are >0.97 and, therefore, much higher than would be expected based on experimental studies (<0.88 ; Dalton & Wood, 1993; Lee & Wyllie, 2000).

The second possibility, fractional crystallization of a carbonate-bearing silicate liquid, resulting in the formation of (exsolved or residual) carbonatitic magmas (Gittins, 1989; Kjarsgaard & Peterson, 1991; Gittins & Jago, 1998; Halama *et al.*, 2005; Brooker & Kjarsgaard, 2011) appears, however, more likely in the case of the KVC. Based on the presence of primary magmatic carbonate in haüyne melilitites, it was concluded that these may represent the 'missing link' between the silicate melts and carbonatites (Keller, 1984a; Hubberten *et al.*, 1988). Liquid immiscibility with evolved haüyne melilitites representing the conjugate silicate melt can be excluded, because a hypothetical unexsolved precursor melt composition (carbonatite+haüyne melilitite) cannot be generated by simple fractionation of the parental olivine melilitite (Supplementary Data Fig. S3). However, since the CO_2 solubility in silicate liquids is highly pressure-dependent (e.g. Yoder, 1975; Mysen *et al.*, 1976), a hypothetical parental melt could have been decarbonated during ascent and hence, the melilititic rocks might not be appropriate for modelling derivation of the carbonatites by differentiation and accompanying CO_2 enrichment. Therefore, we assume that the direct precursor to the carbonatites might be present at lower crustal levels where pressure is high enough to prevent decarbonation reactions. However, whether the carbonatites formed by liquid immiscibility or by extreme magmatic differentiation of a carbonate-rich mafic/ultramafic silicate melt (possibly of olivine melilititic composition) remains speculative.

Mantle source(s) of the KVC rocks

The olivine melilitites of SW Germany formed by partial melting at the base of the lithospheric mantle which previously had been enriched by K-depleted carbonated melts from the dolomite–garnet–peridotite stability field in the asthenosphere (e.g. Dunworth & Wilson, 1998). Ulianov *et al.* (2007) proposed a similar process for the

formation of the olivine/basanitic nephelinites at the KVC and suggested that the more calcic olivine melilitites formed from more strongly metasomatized (carbonated) mantle regions or from smaller degrees of partial melting. However, the lower concentrations of MgO and compatible elements such as Ni and Cr in olivine/basanitic nephelinites compared to olivine melilitites favour their formation by fractional crystallization rather than by the processes mentioned above. The osmium isotope systematics of Blusztajn & Hegner (2002) confirm the lithospheric mantle as the source region, but cannot exclude an asthenospheric contribution.

Monchiquites probably represent somewhat evolved magmas, as indicated by their low MgO, Ni and Cr contents (Figs 8 and 9). Their potassic character cannot be explained by fractionation from olivine melilitites and requires another source. For these rocks, an enriched lithospheric mantle, presumably a phlogopite- and amphibole-bearing spinel lherzolite, was assumed (Wilson & Downes, 1992). Indeed, phonolitic tephrite HTAC1117 contains MgO-rich mica (xeno)crysts which resemble those found in lithospheric mantle xenoliths in the Massif Central and the Rhenish Massif (Wilson & Downes, 1991). Whole-rock isotope data from Schleicher *et al.* (1990) strengthen the assumption of two magma sources, as the olivine/basanitic nephelinites exhibit lower $^{87}\text{Sr}/^{86}\text{Sr}$ isotope ratios ($^{87}\text{Sr}/^{86}\text{Sr} = 0.7032\text{--}0.7040$) than the monchiquites ($^{87}\text{Sr}/^{86}\text{Sr} = 0.7039\text{--}0.7051$).

In summary, at least, two types of parental magmas are necessary to explain the lithological variability of the silicate rocks of the KVC. We envisage that these magmas were produced from a mineralogically and isotopically heterogeneous mantle source, similar to what was proposed for comparable silicate rock-carbonatite complexes at Mt. Vulture (Melluso *et al.*, 1996) or Tamazeght (Marks *et al.*, 2008a).

Redox conditions in carbonatite-producing mantle domains

The few carbonatites for which redox estimates are available (Oka, Chernigovskiy, Jacupiranga, Palabora; see above) contain baddeleyite, perovskite, monticellite or melilite, which suggests $a\text{SiO}_2$ values well below 0.2. Hence, silica-corrected redox conditions are actually several log units above the FMQ-buffer, very similar to our results for the KVC carbonatites (see above; Fig. 18c), and imply highly oxidized parental source regions for carbonatites.

One of the most oxidized mantle regions known are rift-related environments that can reach oxygen fugacities of up to $\Delta\text{FMQ} = +3.5$ (Foley, 2011 and references therein) even without considering the silica-corrected FMQ-buffer (see above). Indeed, many carbonatite complexes, including the KVC, formed in rift settings, with the highest mantle redox estimates being determined for the Olmani peridotites derived from the lithospheric mantle below the East African Rift. Here, rising

carbonate-bearing melts from the asthenosphere caused intense metasomatism (Rudnick *et al.*, 1993, 1994) that probably increased the oxidation state of the lithospheric mantle. Similar processes at the base of the mantle lithosphere are assumed for the source region of the olivine melilitites associated with the KVC (e.g. Dunworth & Wilson, 1998; see above). However, their apparently variable redox conditions (Dunworth & Wilson, 1998) may be caused by contrasting redox states at the boundary between lithospheric and asthenospheric mantle (e.g. Ballhaus, 1993; Ballhaus & Frost, 1994). These differences may have determined the major fractionating phases (olivine versus melilite) during further evolution of the olivine melilititic melts, leading to either nephelinitic–limburgitic or to melilititic–hauynitic rock series (see above). Since the parental liquids to carbonatites are generally assumed to originate from the sub-lithospheric mantle, the source region for those rocks must be heavily influenced by asthenospheric fluids or melts (Bell & Simonetti, 2010). Therefore, the high redox conditions for the rocks of the KVC reflect the interaction between the mantle lithosphere and carbonate-bearing melts or metasomatic fluids emanating from the asthenosphere. The presence of CO_2 during peridotite melting produces silica-undersaturated melts of carbonatitic and melilititic to nephelinitic composition (e.g. Wallace & Green, 1988; Dasgupta *et al.*, 2007). Such associations are found at many carbonatite complexes worldwide (e.g. Kaiserstuhl, Mt. Vulture, Magnet Cove, Tamazeght, Oldoinyo Lengai, Kerimasi) and the silicate rocks often contain mineral assemblages that indicate oxidized crystallization conditions (e.g. perovskite-magnetite-clinopyroxene or garnet-magnetite-wollastonite-clinopyroxene), even where carbonatites are lacking (e.g. Sadiman; Zaitsev *et al.*, 2012).

The oxidation of carbon stored as diamonds at the base of the lithosphere has been proposed as a process to generate carbonate-bearing magmas (carbonate redox melting; Foley *et al.*, 2009; Foley, 2011). As the oxygen fugacity of the mantle increases with decreasing pressure (Ballhaus & Frost, 1994; Woodland & Koch, 2003; Frost & McCammon, 2008), carbonatitic melts may be produced by lithospheric thinning, and hence are often associated with continental rifting. However, present and past subduction zones may equally be responsible for the formation of heterogeneously oxidized mantle domains that may be reactivated much later in an entirely different tectonic setting (e.g. Parkinson & Arculus, 1999; Frost & McCammon, 2008; Martin *et al.*, 2017), as suggested by subduction-like isotope signatures in some carbonatites (e.g. van Achterbergh *et al.*, 2002; Walter *et al.*, 2008; Hulett *et al.*, 2016).

SUMMARY AND CONCLUSIONS

Based on the estimated crystallization parameters (T , $f\text{O}_2$ and $a\text{SiO}_2$) and fractional crystallization models, we propose that two different magma series are responsible for

the various rock types exposed at the KVC. The generally low Na/K ratio of the tephritic to phonolitic rock series implies a K-basanitic parental magma, as previously suggested by Keller (1984a, 1984b) and Kim (1985). The geochemistry and high formation temperatures of the monchiquites (above 1100°C) exclude a genetic link to the more primitive olivine/basanitic nephelinites and limburgites. The evolution of the tephritic–phonolitic rock series via fractional crystallization is characterized by decreasing silica activity and increasing oxygen fugacity, accompanied by mineralogical changes from olivine–clinopyroxene–spinel assemblages in the more primitive rocks via clinopyroxene–magnetite assemblages in the intermediate members to clinopyroxene–garnet–titanite assemblages in the most evolved lithologies. The parental magma of the nephelinitic, melilitic and haüynitic rocks, however, may be similar in composition to olivine melilitites from SW Germany. Variations in the redox state during differentiation may cause variable degrees of melilite versus olivine fractionation and could result in the formation of either olivine/basanitic nephelinites (ΔFMQ^* of around +3) or haüyne melilitites ($\Delta\text{FMQ}^* = +4$ –6). Based on various geochemical arguments and their similarly high redox conditions ($\Delta\text{FMQ}^* = +6$ –7), the carbonatites probably have a genetic relation to the haüyne melilitites.

The highly oxidized conditions of these melts seem to be typical of such complexes as the association of carbonatites with oxidized alkaline silicate rocks is a common feature worldwide. Therefore, we conclude that oxidizing and carbonated fluids/melts interacting with the mantle lithosphere is a major prerequisite for the subsequent genesis of mixed alkaline silicate rock–carbonatite complexes.

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SUPPLEMENTARY DATA

Supplementary data are available at *Journal of Petrology* online.

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Appendix IV

Accepted publication

Study 4

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Pyrochlore as a monitor for magmatic and hydrothermal processes in carbonatites from the Kaiserstuhl volcanic complex (SW Germany)

B.F. Walter^{a,*}, A. Parsapoor^a, S. Braunger^a, M.A.W. Marks^a, T. Wenzel^a, M. Martin^b, G. Markl^a

^a Eberhard Karls Universität Tübingen, Wilhelmstr. 56, D-72074 Tübingen, Germany

^b Landesamt für Geologie, Rohstoffe und Bergbau, Baden-Württemberg, Albertstraße 5, D-79104 Freiburg i. Br., Germany

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ABSTRACT

Pyrochlore from the Kaiserstuhl volcanic complex (SW Germany) shows textural and compositional differences between various coarse-grained calcite-carbonatite bodies (Badberg, Degenmatt, Haselschacher Buck, Orberg) and extrusive carbonatites (Henkenberg, Kirchberg).

Oscillatory-zoned F-rich pyrochlore with up to 69 wt% Nb₂O₅ is common in all coarse-grained calcite-carbonatite bodies and probably formed during magmatic conditions. However, only in some of the samples from the Badberg, partly resorbed U- and Ta-enriched pyrochlore cores with up to 22 wt% UO₂ and 9 wt% Ta₂O₅ have been identified, which are interpreted as being inherited from underlying nosean syenites. Pyrochlore data from a drill core penetrating the Badberg indicate increasing contents of REE, U, and Ta with depth, while Nb, F and Na contents decrease. This may reflect the combined effects of fractional crystallization and assimilation (AFC) or indicates a multi-stage emplacement of the carbonatitic magma. Patchy-zoned ceriopyrochlore and REE- and Th-enriched pyrochlore with up to 19 wt% total REE₂O₃ and 6.5 wt% ThO₂ is largely restricted to samples from the Orberg and probably formed during hydrothermal conditions. This can be related to the relatively evolved character of the Orberg carbonatites, based on their relatively high whole-rock Nb/Ta and Zr/Hf mass ratios. This study demonstrates that the textural and compositional variation of pyrochlore in carbonatites is a powerful tool to distinguish magmatic, hydrothermal and weathering processes in carbonatitic systems.

1. Introduction

The so-called High Field Strength Elements (HFSE) include the rare earth elements (REE), yttrium (Y), zirconium (Zr), hafnium (Hf), niobium (Nb), and tantalum (Ta) that are essential for many high-tech applications, such as semiconductors, displays, batteries and high performance magnets (e.g., Humphries, 2010 and references therein). The industrial demand for this element suite largely increased during the last decades and recently, the European Commission recognized a significant risk of supply for some of the HFSE and defined them as critical metals (e.g., Du and Graedel, 2011; Hatch, 2012; Massari and Ruberti, 2013; EC, 2014).

Deposits for HFSE are mostly found in alkaline rocks and carbonatites, the latter being recently defined as rocks with ≥ 30 vol% primary igneous carbonate minerals (Mitchell, 2005). Most carbonatites are associated with various types of alkaline silicate rocks, ranging from ultramafic (e.g., olivinites and pyroxenites in Kovdor; Krasnova et al., 2004a) to felsic (e.g., nepheline syenites at Grønnedal-Ika, Greenland; Halama et al., 2005). Among the > 500 known occurrences, only 49

contain extrusive carbonatites (Woolley and Church, 2005; Woolley and Kjarsgaard, 2008).

The most common HFSE-carriers in carbonatites are pyrochlore-group minerals, perovskite and titanite-group minerals, baddeleyite (ZrO₂), zirconolite (CaZrTi₂O₇), calzirtite (Ca(Zr,Ca)₂Zr₄(Ti,Fe)₂O₁₆), zircon and apatite (e.g., Chakhmouradian, 2006). Pyrochlore-group minerals are especially abundant in carbonatites and are a major Nb source to date (Mitchell, 2015). Three mines are in operation today: Araxá and Catalão-II (Brasil) and St. Honoré (Canada). Numerous occurrences are under evaluation (e.g., Aley, Upper Fir in Canada, Tomtor in Russia, Lueshe in Dem. Rep. Congo) or re-evaluation (e.g., Panda Hill in Tanzania, Oka in Canada and Fen in Norway; Gendron et al., 1981; Issa Filho et al., 1984; Mariano, 1989, Lapin and Tolstov, 1995; Nasraoui and Bilal, 2000; Mitchell, 2015 and references therein). Some carbonatites are mined for REEs (e.g., Mount Weld in Australia; Humphries, 2010, Mountain Pass in USA; Castor, 2008; and Bayan Obo in Mongolia; Yang et al., 2011) and one occurrence (Kovdor in Russia; Ivanyuk et al., 2002; Krasnova et al., 2004b) is mined for Zr with baddeleyite being an ore mineral.

* Corresponding author.

E-mail address: benjamin.walter@uni-tuebingen.de (B.F. Walter).

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The Kaiserstuhl volcanic complex (SW Germany) consists of a variety of alkaline silicate rocks associated with intrusive and extrusive carbonatites (e.g., [Wimmenauer, 2003](#), [Braunger et al., in press](#)). This eroded volcanic complex (even though it is a subeconomic occurrence) was chosen by the HiTech AlkCarb project as a natural laboratory for studying the textural, mineralogical and geochemical relationships between intrusive and extrusive carbonatites and to investigate depth-dependent systematics, which could help to improve existing geomodels (e.g., [Le Bas, 1987](#)) and exploration methods for carbonatites. We focus here on pyrochlore-group minerals because this mineral group shows significant compositional variations in various magmatic, hydrothermal and weathering stages (e.g., [Wall et al., 1996](#); [Chakhmouradian, 2006](#); [Lumpkin and Ewing, 1995](#); [Williams et al., 1997](#); [Zaitsev et al., 2012](#)). Hence, the textural and compositional variability of pyrochlore-group minerals provides the opportunity to study the geochemical evolution of a carbonatite system in detail.

Previous studies indicate mineralogical and textural variations within and between the different carbonatite bodies of the complex ([Blust, 1993](#); [Katz and Keller, 1981](#); [Keller, 1964, 1965, 1981, 1984a, 1984b, 1989](#); [Keller et al., 1990](#); [Sigmund, 1996](#)). This present study is guided by the following questions that have both economic and fundamental significance:

- (1) Are there systematic textural and compositional variations of the pyrochlore-group minerals between the different carbonatite types of the complex?
- (2) Are there depth-dependent textural and compositional variations of pyrochlore-group minerals in coarse-grained calcite-carbonatites?
- (3) Is it possible to use these variations and differences for improving exploration methods for carbonatite-related HFSE mineralizations?

2. The Kaiserstuhl volcanic complex

The Kaiserstuhl volcanic complex (SW-Germany) is situated in the Upper Rhinegraben rift and is part of the Central European Magmatic Province (e.g., [Wilson and Downes, 2006](#); [Lustrino and Wilson, 2007](#); [Fig. 1A](#)). The complex is situated at the junction of four tectonic blocks and is affected by Neogene strike slip and horst-graben tectonics in the Rhinegraben ([Fig. 1B & C](#); [Beccalotto et al., 2010](#)). The rocks of this partly eroded stratovolcano are of Miocene age (18–15 Ma) and are dominated by mostly tephritic lavas, ash beds, sills, and dykes that are interbedded with rare nephelinitic lavas and extrusive carbonatites ([Fig. 1D](#)). This sequence was intruded by several phonolite bodies and larger masses of sodalite monzogabbros. The latter are essentially confined to the central part of the complex ([Fig. 1D](#); e.g., [Wimmenauer, 2003](#) and references therein). Subsequently, polygenic diatreme breccias formed (e.g., [Baranyi, 1971](#); [Sigmund, 1996](#)) and established pathways for several larger carbonatite bodies ([Fig. 2A](#)). These are in turn intruded by phonolites that are very rarely associated with medium- to coarse-grained nosean syenite (only in the Badberg area; see [Braunger et al., in press](#); [Giebel et al., submitted](#)). The latter rocks are known to occur at depth, based on a scientific (unfortunately lost today) drilling at the Badberg (Forschungsbohrung Kaiserstuhl FB, [Kirchheimer, 1973](#)) and their presence as cm- to dm-sized xenoliths in various lithologies across the Kaiserstuhl ([Wimmenauer, 2003](#); [Seeger, 2015](#)). Hence, nosean syenites underlie at least parts of the complex, as has been postulated by [Goodenough et al. \(2016\)](#). Today, most of the Kaiserstuhl rocks show variable degrees of surficial weathering and are partly covered by several tens of meters of Quaternary fluvial and aeolian sediments (loess).

The magmatic rocks of the Kaiserstuhl have been interpreted to represent two magma series: Sodic rocks (olivine nephelinites, haiyne melilitites and (melilite) haiynites) derived from a primitive magma and a more differentiated magma type that formed sodic-potassic rocks (tephrites, phonolites, e.g., [Schleicher et al., 1990](#), [Wimmenauer, 2003](#)). Based on isotopic studies, [Schleicher et al. \(1990\)](#) found

evidence for a genetic link between the carbonatites and the Na-series magmas of the Kaiserstuhl. The detailed processes that generated the carbonatites, however, are still under debate (e.g., [Wimmenauer, 2003](#); [Wang et al., 2014](#); [Braunger et al., in press](#)).

2.1. Investigated sample material

More than 200 samples from drill cores, surface outcrops and sample collections at the Universität Tübingen and the LGRB (Geological Survey of Baden-Württemberg) were investigated for this study. The complete sample list (including GPS positions) and details on the scientific drillings are given in Electronic supplements ES1 and ES2, respectively. Two of the four drill cores from the Kaiserstuhl (KB2 1991 and KB3 1991) are still available (see [Fig. 2A](#) for their locations). The 195 m deep KB2 drill core from Orberg quarry II (Electronic supplement ES3) consists of coarse-grained carbonatite underlain by polygenic diatreme breccias crosscut by sodalite monzogabbro and carbonatitic veins ([Sigmund, 1996](#)). The 400 m deep KB3 drill hole at the Steinreise location (on a small hill called Horberig immediately west of the Badberg) penetrated sodalite monzogabbros and alkaline dyke rocks and underlying coarse-grained carbonatite with nosean syenite ([Fig. 2A](#), Electronic supplement ES4). The whole profile is crosscut by various carbonatitic and silicate dyke rocks ([Blust, 1993](#)). Not preserved are the cores of the 100 m deep KB1 (1991) in the Eichstetter Tal and the 500 m deep the Badloch drill cores (1970, Electronic supplement ES5). The KB1 drilling (100 m of coarse-grained calcite-carbonatite) failed and the cores of the Badloch drilling were destroyed in a fire. Based on petrographic descriptions from the 1970s ([Kirchheimer, 1973](#)), the Badloch drilling (FB) cored coarse-grained calcite-carbonatite underlain by nosean syenite ([Fig. 2A & B](#)), interrupted by phonolitic rocks. Again, the profile was crosscut by numerous carbonatitic and silicate dyke rocks. All surficial samples (with the exception of those taken from quarry faces) show variable degrees of surficial weathering (e.g., increased porosity, alteration of mica and magnetite to secondary FeOOH-phases).

2.2. Geology of the carbonatites

2.2.1. Coarse-grained calcite-carbonatites

Coarse-grained calcite-carbonatites (sövites) are restricted to the central part of the Kaiserstuhl, where they occur as four larger bodies (Badberg, Haselschacher Buck, Degenmatt, Orberg) and two small satellite pipes (Katharinenberg and Kirchberg, [Figs. 1 & 2](#)). The largest of these bodies (Badberg) exhibits layering structures parallel to its contact towards the country rocks and contains elongated xenoliths of fenitized nosean syenites that are similarly aligned. The abundance of these xenoliths decreases from the upper to the lower parts of the body ([Fig. 2B](#)). A tabular geometry of the Badberg body is suggested, with a NW-dipping of about 60° and an inclination-corrected thickness of < 300 m ([Wimmenauer, 1963](#); [Fig. 2](#)). The hanging wall consists of sodalite monzogabbros, the foot wall is represented by phonolites and nosean syenites and numerous dyke rocks crosscut the Badberg carbonatite ([Kirchheimer, 1973](#); [Fig. 2B](#)). The Degenmatt and Haselschacher Buck occurrences resemble similar structures as the Badberg but their unsatisfactory outcrop situation does not allow for a detailed structural analysis. At the Orberg, coarse-grained calcite-carbonatite was mined in five quarries. Quarries I-III expose an up to 10 m thick NNE-SSW striking dyke-like body. The variable dip of this body (about 70° in quarry III but only about 25° in quarry II; [Wimmenauer, 1963](#)) and the variably oriented layering structures in quarries IV and V ([Hubaux, 1964](#); [Fig. 2](#)) indicate a cauliflower-like structure consisting of several dyke/sill-like apophyses surrounded by polygenic diatreme breccias ([Fig. 2](#)).

The Kaiserstuhl is situated on the junction between four tectonic blocks ([Fig. 2B](#)) and is intersected by the regional Tuniberg fault ([Figs. 1 & 2](#)) with a down-throw towards the graben center and a vertical

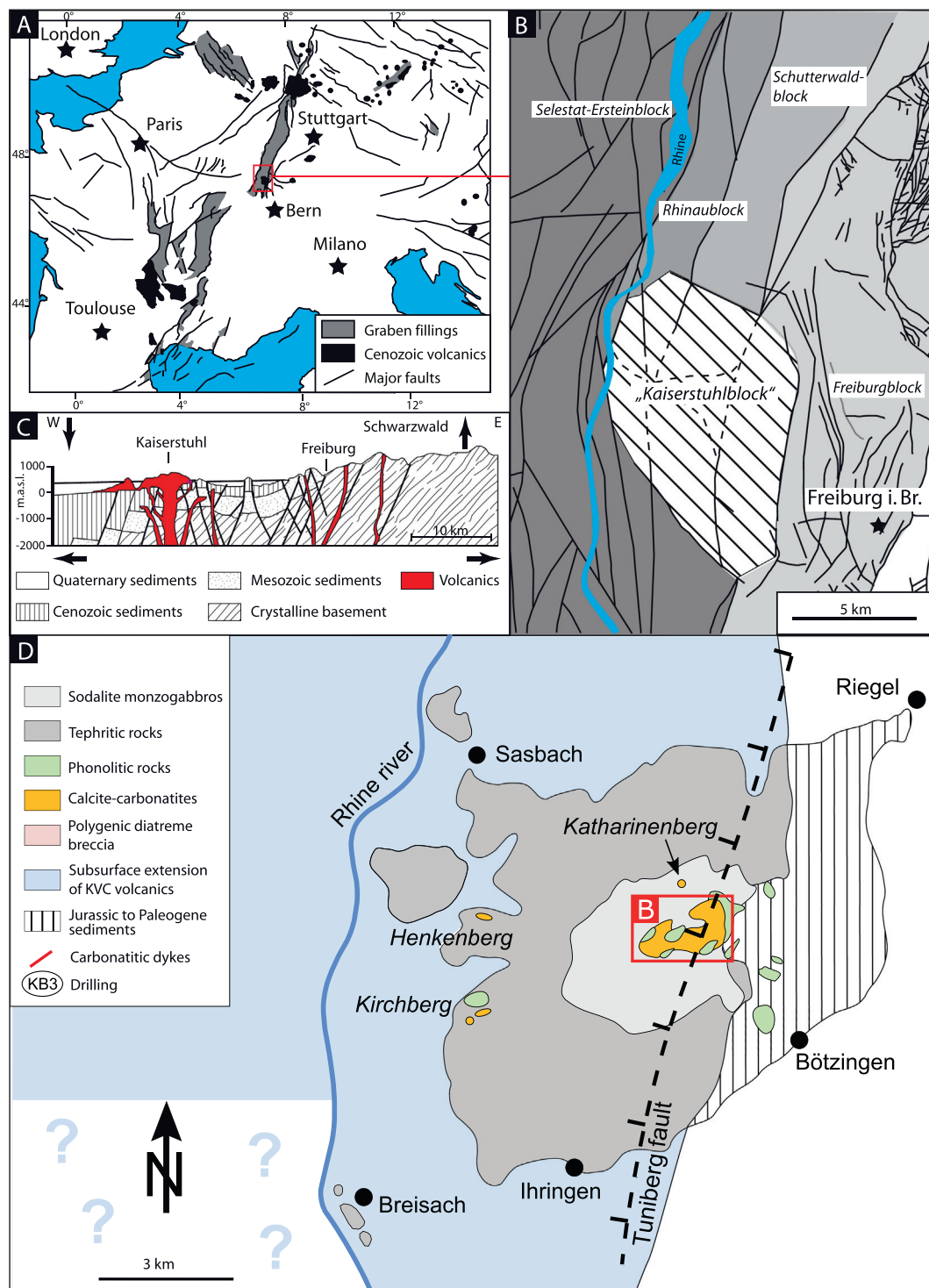


Fig. 1. (A) Overview map of the Central European Magmatic Province including Palaeogene rifts (grey) and volcanic products (black). (B) Subsurface tectonic map of the Kaiserstuhl area after the database of [Beccaletto et al. \(2010\)](#). The Kaiserstuhl is situated on four different tectonic blocks. (C) Cross section of the eastern Upper Rhine graben. Note that the extrusive rocks cover the Miocene surface, which is today mostly covered by Quaternary gravels. (D) Simplified geological map of the Kaiserstuhl volcanic complex (modified after [Keller et al., 1990](#)).

displacement of 1000–3000 m ([Beccaletto et al., 2010](#)). Therefore, we speculate that the Badberg, Degenmatt and Haselschacher Buck occurrences once belonged to a common sub-vertical pipe-like structure that was subsequently tilted and disrupted by Neogene strike-slip and horst-graben tectonics within the Upper Rhinegraben ([Beccaletto et al., 2010](#)). The Orberg occurrences may represent a larger apophysis of this hypothetical structure (e.g., [Hubaux, 1964](#)), whereas the small (< 10 m

in diameter) and isolated occurrences at Kirchberg and Katharinenberg are interpreted as two minor pipe-like structures with unknown relationships to the main structure ([Fig. 2](#)).

2.2.2. Medium- to fine-grained carbonatites

Medium- to fine-grained calcite-carbonatites (alvikites) occur as (sub)vertical dykes with variable thickness (cm to m). Most of them are

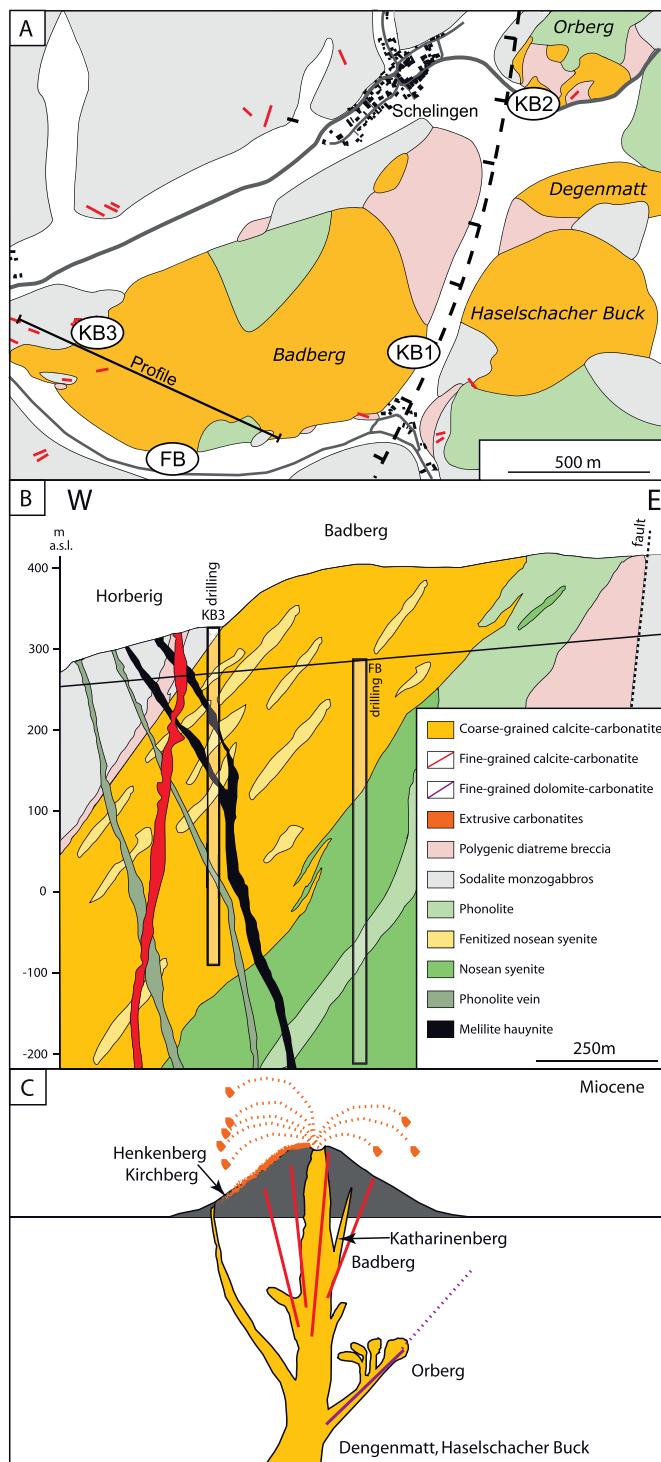


Fig. 2. Geology of the Karbonatites in the Kaiserstuhl. (A) Simplified map of the central part of the Kaiserstuhl (modified from [Wimmenauer, 2003](#)) with locations of the four scientific drillings. (B) Schematic W-E profile through the Horberig and western Badberg area. The profile is based on field observations and drill cores and logs. The black subhorizontal line indicates the valley S of the Badberg (see [Fig. 2A](#)). (C) Schematic illustration (not to scale) of the hypothetical, initially (sub)vertical, pipe-like structure showing the different carbonatite occurrences at the Kaiserstuhl.

disclosed in the central part of the complex, but few occur in the outer extrusive tephrites. They crosscut all other rock types ([Fig. 2](#)) and represent the youngest volcanic rocks of the complex ([Katz and Keller, 1981](#)). Most of them are assumed to belong to a radial dyke (cone sheet)

system with its center below the Badberg ([Lehnert, 1989](#)).

Medium- to fine-grained dolomite-carbonatites (beforsites) rarely occur as cm- to m-sized dykes, apophyses, and irregularly formed bodies within coarse-grained calcite-carbonatites, close to the contact with the (Mg- and Fe-rich) host rocks. They occur in Orberg quarries II, III and V and in a previously not described occurrence at Degenmatt.

2.2.3. Extrusive carbonatites

At the western flank of the complex, up to three layers (each 1–1.5 m thick) of extrusive carbonatites occur at Henkenberg and Kirchberg ([Fig. 1D](#)). They represent alternating sequences of carbonatitic crystal tuffs, lapillistones and lavas interbedded with tephritic and basanitic ashes and lavas ([Keller, 1981, 1989](#); [Woolley and Kjarsgaard, 2008](#)). The distance between these two localities is about 1.5 km but because of intense anthropogenic land transformations in the 1970s only isolated outcrops are preserved and their original extent is unknown.

3. Petrography

3.1. Coarse-grained calcite-carbonatites

Detailed petrographic descriptions of these rocks are already available (e.g., [Wimmenauer, 1963, 2003](#); [Wimmenauer et al., 1977](#); [Lehnert, 1989](#); [Keller et al., 1990](#)). These are briefly summarized here and we provide details on the observed textures for pyrochlore-group minerals, as these are the focus of the present study. Coarse-grained calcite-carbonatites are dominated by calcite and contain variable amounts of apatite, magnetite, phlogopite, pyrochlore-group minerals and rarely forsteritic olivine, monticellite and melilite. Common accessories include sulfides (pyrite, pyrrhotite, galena, sphalerite, chalcocopyrite), barite and rare Nb-perovskite (only in some Badberg samples). The modal amount of non-carbonate minerals varies largely on the dm- to m-scale and in some areas mineral banding occurs, represented by alternating coarse-grained and finer-grained layers with very different modal amounts of apatite, magnetite and pyrochlore-group minerals.

Pyrochlore-group minerals occur in all investigated samples. They form up to 2 mm large anhedral to euhedral grains, in cases intergrown with apatite and sulfides. In samples from Badberg, Haselschacher Buck and Degenmatt, pyrochlore-group minerals typically show oscillatory zoning patterns ([Fig. 3A–F](#)). In Badberg samples taken in the vicinity of fenitized nosean syenites and from deeper levels of the KB3 drill core (see above), however, some pyrochlores contain BSE-bright cores that are partly resorbed and surrounded by the before-mentioned oscillatory-zoned mostly BSE-darker areas ([Fig. 3D](#)). Such cores were also recognized at Degenmatt and Haselschacher Buck ([Fig. 3E & F](#)). However, the bad outcrop situation at these locations does not allow to judge if these are related to fenitized nosean syenite xenoliths. At Orberg and Katharinenberg however, such bright cores (and xenoliths of fenitized nosean syenite) are typically missing ([Fig. 4A–H](#)) and pyrochlore group minerals are mostly oscillatory zoned ([Fig. 4A–H](#)) and contain late-stage BSE-bright rims ([Fig. 4B](#)) or show patchy zonation (with BSE-brighter alteration patches) textures that overprint the primary oscillatory zoning ([Fig. 4F](#)). Typically, the patchy overprinting starts at the grain boundary whereby an alteration front moves towards the core area of the pyrochlore ([Fig. 4F](#)). Furthermore, the common association with rosettes of apatite needles indicates that at least the pyrochlore group minerals in the banded carbonatites of the complex crystallized in-situ ([Fig. 4C & D](#)).

3.1.1. Fenitized nosean syenites

At the Badberg, fenitized nosean syenites occur as up to several meters large xenoliths in the coarse-grained calcite-carbonatite (see above; [Fig. 2B](#)). These rocks show a primary mineralogy of hauyne \pm alkali feldspar \pm clinopyroxene \pm garnet \pm apatite \pm magnetite \pm titanite \pm pyrochlore-group minerals that is variably metasomatized.

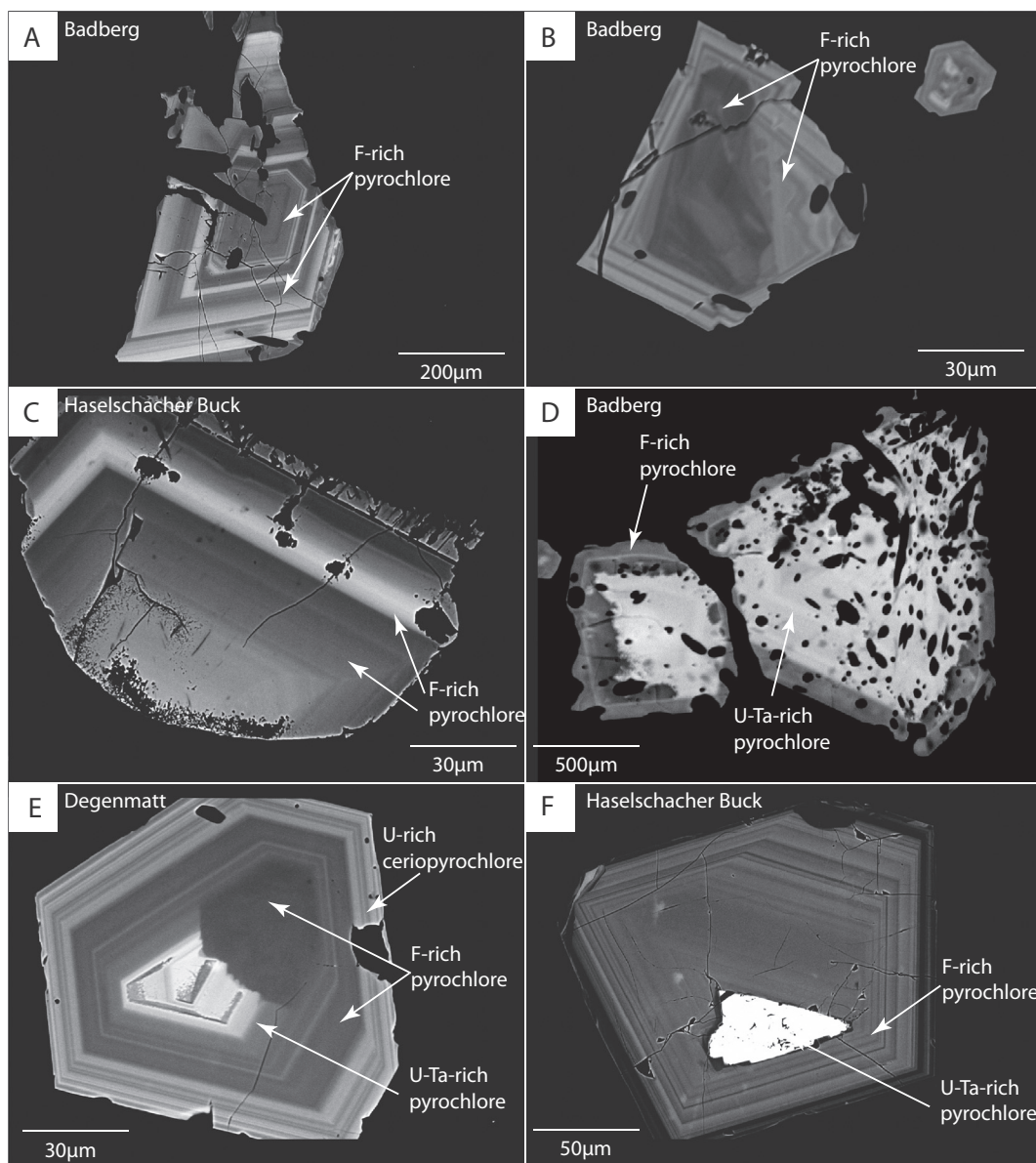


Fig. 3. Pyrochlore textures from coarse-grained calcite-carbonatites of the Kaiserstuhl (Badberg, Haselschacher Buck, Degenmatt). (A) & (B) Oscillatory zoned. F-rich pyrochlore from Badberg. (C) Oscillatory zoned F-rich pyrochlore from Haselschacher Buck. (D) Pyrochlore from Badberg with typical U-Ta-enriched cores overgrown by oscillatory zoned F-rich pyrochlore. (E) Oscillatory zoned pyrochlore with a BSE-bright core from Degenmatt. (F) Oscillatory zoned F-rich with BSE-bright U-Ta-enriched core from Haelschacher Buck.

The samples contain variable amounts of coarse-grained interstitial calcite and former nosean or haüyne and alkali feldspar are largely now decomposed to various zeolite minerals and calcite. Clinopyroxene occurs mostly as pale green, sometimes patchy-colored subhedral and interstitial grains and if present, a dark brown garnet I is overgrown by a reddish-brownish garnet II that occasionally forms poikilitic aggregates that enclose former foid minerals (for details see Giebel et al., submitted). Pyrochlore-group minerals typically consist of BSE-bright cores that are partly resorbed and overgrown by oscillatory-zoned areas.

3.2. Medium- to fine-grained carbonatitic dykes

These rocks include banded, comb-layered, porphyritic and streaky varieties of medium- to fine-grained calcite-carbonatites described in detail in Katz and Keller (1981), Hubberten et al. (1988) and Lehnert (1989). Briefly, they consist of a groundmass of calcite with bands or striations of magnetite, apatite, pyrochlore (mostly unzoned) and Nb-

perovskite. In some cases, calcite ocelli, phlogopite, rare alabandine and primary barite were recognized. Porphyritic varieties further contain rounded and deformed clasts of coarse-grained calcite and accessory garnet, pyrochlore, magnetite, amphibole, forsteritic olivine, zirconolite, calzirite and baddeleyite. Rare medium- to fine-grained dolomite-carbonatites consist of a fine-grained groundmass of dolomite-ankerite and contain the same non-carbonate minerals as medium- to fine-grained calcite-carbonatites, with additional bastnaesite and monazite (Wimmenauer, 2003).

3.3. Extrusive carbonatites

The petrography of the extrusive carbonatites from the complex has been described in detail before (e.g., Keller, 1981, 1989; Hubberten et al., 1988). Briefly, carbonatitic crystal tufts from Henkenberg are mainly composed of sparry calcite with clasts of fenitized nosean syenites, coarse-grained calcite, magnetite, garnet and augitic pyroxene,

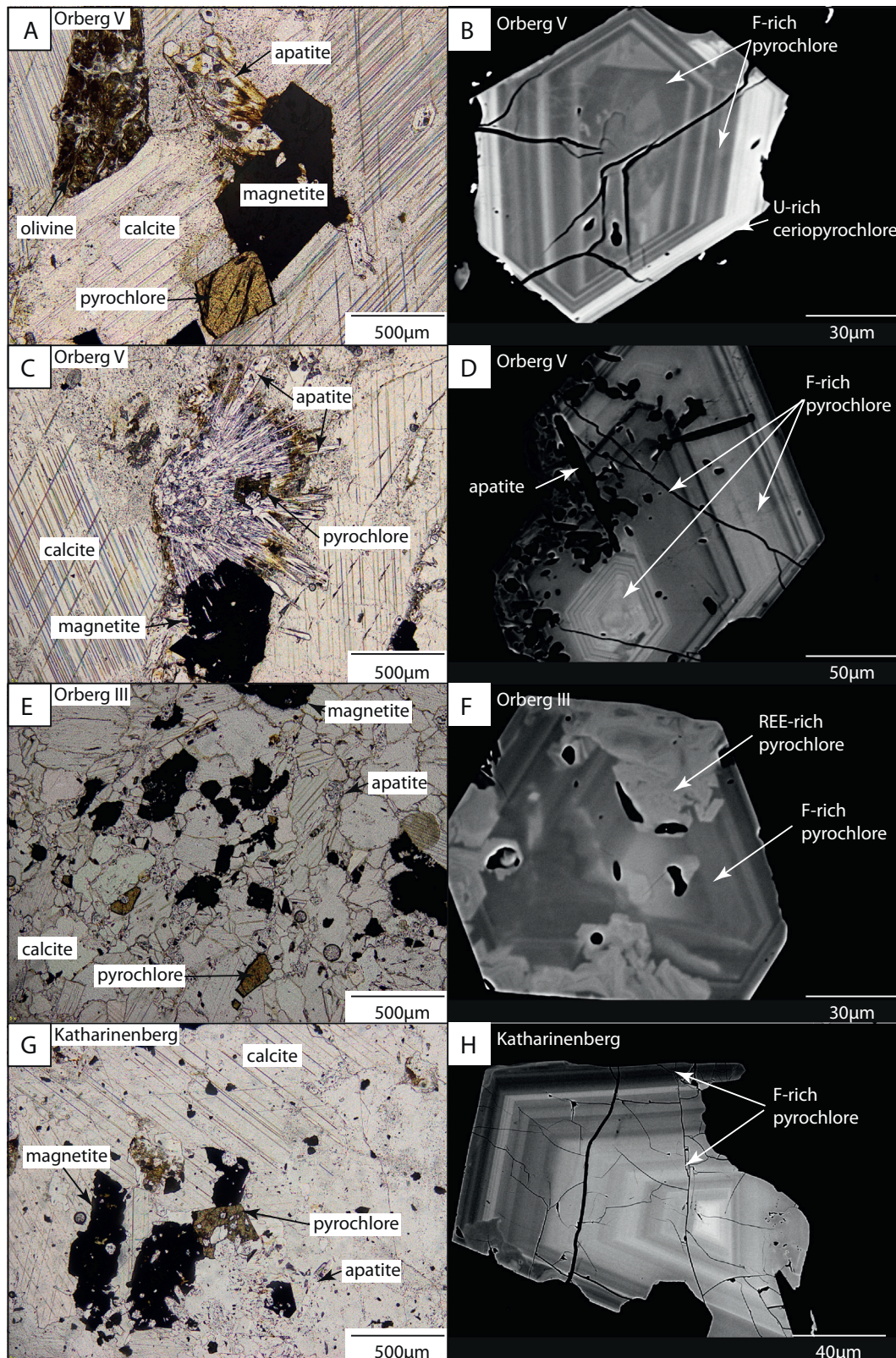


Fig. 4. Pyrochlore textures from coarse-grained calcite-carbonatites of the Kaiserstuhl (Orberg, Katharinenberg). (A) Euhedral pyrochlore associated with magnetite and apatite from Orberg quarry V. (B) Typically, at Orberg the U-Ta-enriched pyrochlore cores are missing. Often the oscillatory zoned Orberg pyrochlores contain U-rich ceriopyrochlore or U-REE-Th enriched pyrochlores rims. (C & D) At Orberg quarry V and elsewhere, rosettes of apatite needles are over- and intergrown with pyrochlore. (E & F) In the finer-grained calcite-carbonatite of Orberg quarry III pyrochlores often occur as euhedral grains with a patchy zonation that overprints the primary oscillatory zoning. (G & H). Pyrochlore from Katharinenberg is typically oscillatory zoned and lacks U-Ta-enriched BSE-bright cores.

and rare apatite and pyrochlore-group minerals. Carbonatitic lapilli tuffs consist of closely-packed lapilli of 0.5 to 10 mm diameter, cemented by secondary calcite. The lapilli have a porphyritic and trachtyoid texture with phenocrysts of platy calcite, minor magnetite and accessory clinopyroxene, garnet, apatite, Nb-perovskite, and pyrochlore-group minerals. A carbonatite lava from Kirchberg consists of a fine-grained and altered carbonate-rich matrix with platy calcite phenocrysts accompanied by minor euhedral to subhedral magnetite, garnet, nepheline, and alkali feldspar.

4. Analytical methods

4.1. Whole-rock analyses

Prior to analysis, all samples (generally ≥ 1 kg) were liberated from weathering crusts and crushed. About 50 g representative sample aliquots were then ground with an agate mill for 10 min. Loss on ignition was determined from 1 g sample material at 1000 °C.

Major elements were analysed by means of wavelength dispersive XRF at Universität Tübingen (Germany). For the fused beads 1.5000 g of dried sample powder (at 105 °C) was mixed with 7.5000 g MERCK spectromelt A12 (mixture of 66% Li-tetraborate and 34% Li-metaborate) and melted at 1200 °C beads using an Oxiflux system from CBR analytical service. Measurements were done using a Bruker AXS S4 Pioneer system (Rh-tube at 4 kW) with 32 standardised samples (Govindaraju, 1989). Analytical uncertainties are usually in the range of 0.1% (relative), and detection limits are in the range of 1–10 $\mu\text{g/g}$.

Trace element contents were determined with ICP-MS (X-series 2, Thermo Fisher Scientific) at the Karlsruhe Institute of Technology (Germany) using acid digestion of 100 mg of representative sample powder. To assure complete silicate decomposition, 40% HF (suprapur), 65% HClO₄ (normatom) and the pre-oxidized (65% HNO₃, subboiled) sample were heated in a closed Teflon vessel for 16 h at 120 °C. After evaporating the acids to incipient dryness, the residue was re-dissolved in 65% HNO₃ (subboiled) and evaporated again (three times) for purification purposes. The final residue was dissolved in 50 ml of ultrapure water. The quality assurance of the trace element ICP-MS measurements was done by including the certified reference material CRM-TMDW-A (High-Purity standards, Inc.) into the protocol (accuracy: $\pm 10\%$ for most elements). Mathematical corrections of Eu, Gd, Tb and Er were necessary due to high concentrations of Ba and light REE. ^{151/153}Eu was corrected for BaO interferences, ¹⁵⁷Gd for PrO and ¹⁵⁸Gd for CeO interferences, ¹⁵⁹Tb and ¹⁶⁷Er for NdO interferences. The mean value of both isotopes is given for Eu and Gd. To assure the quality of the whole procedure, six blanks and two certified reference materials (SY-2, SY-3; Govindaraju, 1994) were included into the digestions process (accuracy: mainly $\pm 10\%$). The reproducibility ($\pm 5\%$ for most elements) was checked by digesting a phonolitic sample in triplicate.

4.2. Electron microprobe analyses

For analyses of pyrochlore-group minerals a JEOL 8900 electron microprobe at the University of Tübingen was used with a beam current of 10 nA and an acceleration voltage of 20 kV in connection with a focused beam. For calibration, diopside was used for Ca and Si, albite for Na, hematite for Fe, bustamite for Mn, ThO₂ for Th, SrTiO₃ for Sr and Ti, Li₂Ta₂O₆ for Ta, Al₂O₃ for Al, LiNb₂O₆ for Nb, ZrO₂ for Zr, YAG-glass for Y, UO₂ for U, La-glass (REE16G) for La, Ce-glass (REE16G) for Ce, Pr-glass for Pr (REE16G), Sm-glass for Sm (REE16G), and CaF₂ was used for F. Counting times on the peak were 16 s for Ca, 30 s for Na, Si, Sr, Nb, Ta, Al, Y, Fe, Mn, La, Ce, Sm, Pr, Nd, Ti, Zr, U, Th and Sr and 60 s for F. Background counting times (upper and lower) were half the peak times. Data reduction was performed using the internal ZAF matrix correction software of JEOL (Bence and Albee, 1968; Armstrong, 1988, 1991). Details on the WDS configuration used and typical detection limits for the individual elements are given in the Electronic

supplement ES6. High-precision quantitative analyses of F are difficult, as for light elements, significant peak shifts may occur that result in unreliable net peak intensities (e.g., Bastin et al., 1986; Wenzel et al., 2003). Therefore, the area intensity measurement mode was applied for determining F contents. Hereby, the WDX signal was step-wise integrated from the lower to the upper background along the F peak position on a F-rich pyrochlore of sample HTAC_1200.

5. Results

5.1. Whole rock data

All investigated samples (N = 43) are classified as calcio-carbonatites (Woolley and Kempe, 1989), as they are low in Na₂O (< 1.4 wt%), Fe₂O₃ (< 8.4 wt%), and MgO (< 3.7 wt%), with no clear differences between intrusive and extrusive samples or localities (Electronic supplement ES7). As being typical for carbonatites, primitive-mantle normalized trace element patterns (Fig. 5A) show variable enrichment of LILE (e.g., Ba, Sr) and P, with some HFSE being strongly enriched (Nb, Ta, REE) and others showing variable depletion (Zr, Hf, U, Th, Ti). Compared to global carbonatite data (Chakhmouradian, 2006), the contents of Nb (up to 2700 $\mu\text{g/g}$) are relatively high, whereas Zr contents (< 300 $\mu\text{g/g}$) are rather low (Fig. 5B & C). All investigated samples have typical LREE-enriched patterns and the absolute REE contents (up to about 2700 $\mu\text{g/g}$) roughly correlate with P₂O₅ (Fig. 5D).

5.2. Compositional variation of pyrochlore-group minerals

The general formula of this mineral group is A_{2–m}B₂X_{6–w}Y_{1–n}pH₂O (Lumpkin and Ewing, 1995; Atencio et al., 2010). At the A site, As, Ba, Bi, Ca, Cs, K, Mg, Mn, Na, Pb, REE, Sb, Sr, Th, U and Y are incorporated, the B site is normally occupied by Nb, Ta, Ti, Zr, Fe³⁺, Al and Si, and the Y and X positions are filled with O, OH and F (Zurevinski and Mitchell, 2004). Vacancies are very common in the A and Y sites and increase with low-temperature weathering (Lumpkin and Ewing, 1995; Wall et al., 1996). In all, > 1200 EPMA analyses of pyrochlore-group minerals were carried out (Electronic supplement ES8), representative analyses are given in Table 1. The formulae have been calculated on the basis of 2 B-site cations and all analyses fall into the pyrochlore-subgroup (Fig. 6) following the classification of Hogarth (1977). The different pyrochlore types (resorbed cores, oscillatory zoned and patchy zoned; see above) show clear compositional differences (Fig. 7 and Electronic supplements ES9–ES11):

- (i) The BSE-bright cores restricted to pyrochlores from Badberg, Degenmatt and Haselschacher Buck (Fig. 4D–F) are low in Na₂O (≤ 4.1 wt%). They are enriched in UO₂ (up to 22.2 wt%) and Ta₂O₅ (up to 9.3 wt%), but have low ZrO₂ (≤ 0.1 wt%), ThO₂ (b.d.l.) and F (≤ 0.5 wt%), and moderate REE oxide (up to 7.1 wt%) contents.
- (ii) Oscillatory-zoned pyrochlore occurs in all investigated samples (Fig. 3A–F; 5A–H) and is rich in CaO (up 23.2 wt%) and Na₂O (up to 11.3 wt%). Compared to the above-mentioned BSE-bright cores, they are much lower in UO₂ (≤ 4.3 wt%), and Ta₂O₅ (≤ 1.9 wt%), but have similar REE oxide (up to 12.0 wt%) contents and reach higher contents of ZrO₂ (up to 5.8 wt%), ThO₂ (up to 1.4 wt%) and F (up to 3.4 wt%).
- (iii) Patchy-zoned pyrochlore (Fig. 4F) from Orberg is relatively low in CaO (≤ 16.3 wt%) and Na₂O (≤ 5.9 wt%), while UO₂ (up to 8.2 wt%), Ta₂O₅ (up to 4.8 wt%), ThO₂ (up to 6.5 wt%), ZrO₂ (up to 13.7 wt%) and the REE oxides total (up to 18.4 wt%) reach higher contents than in oscillatory-zoned pyrochlore. Some analyses indicate TREES > 0.4 apfu and fulfill the criteria for the terminus ceriopyrochlore (Hogarth, 1977).

Pyrochlore from medium- to fine-grained calcite-carbonatites is generally unzoned but the compositional variations among grains of the

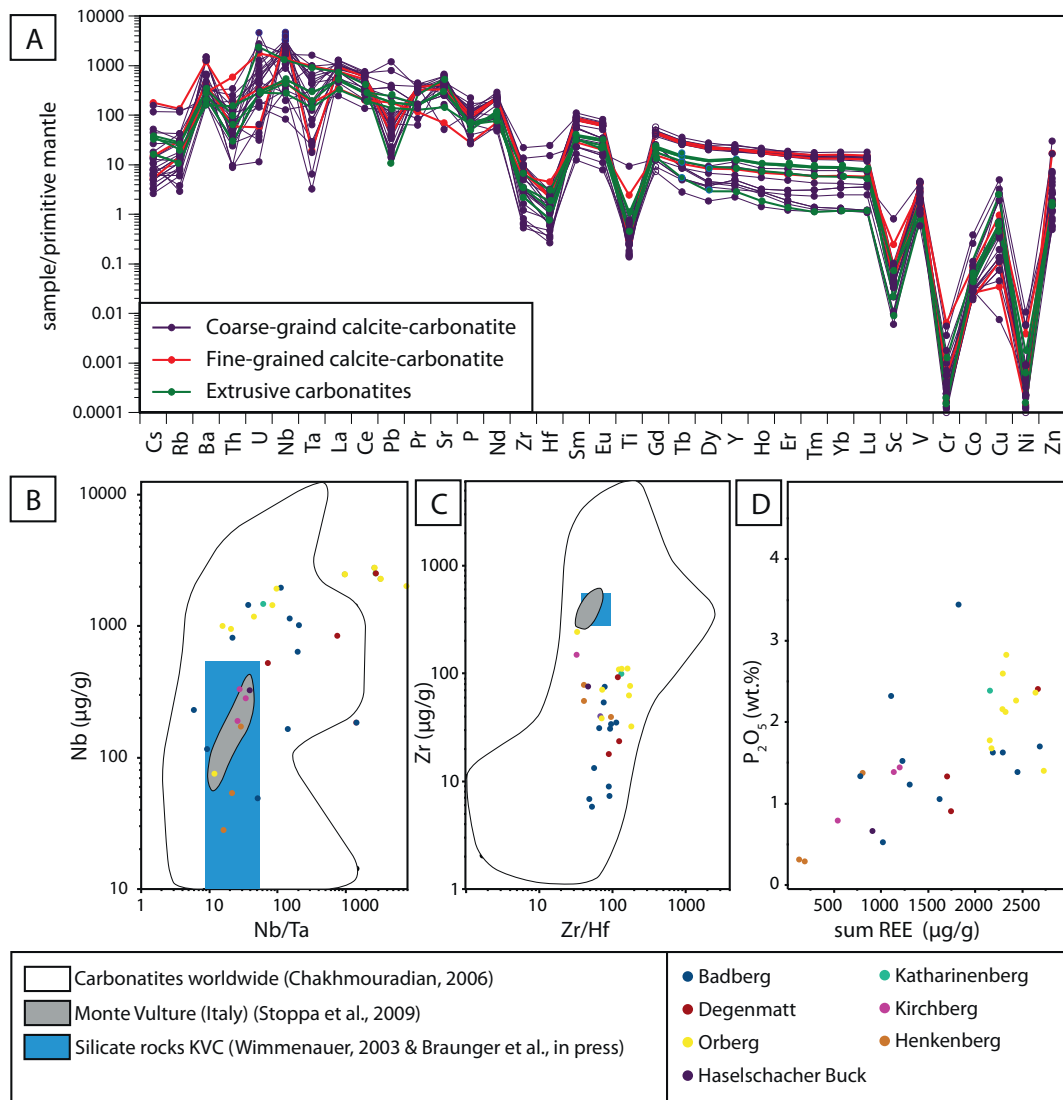
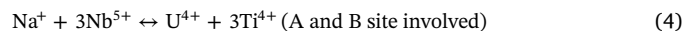
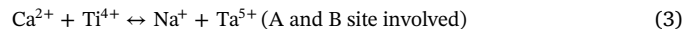
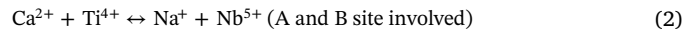


Fig. 5. Whole-rock data for Kaiserstuhl carbonatites. (A) Primitive mantle-normalized trace element patterns. (B) & (C) Plots of Nb and Zr concentration versus Nb/Ta and Zr/Hf ratios of the Kaiserstuhl carbonatites compared to global carbonatite data (Chakhmouradian, 2006); data for carbonatites from Mt. Vulture (Rosatelli et al., 2007) and data for Kaiserstuhl silicate rocks (Braunger et al., in press). (D) Correlation between whole-rock P₂O₅ (wt%) and sum of REEs (µg/g) for Kaiserstuhl carbonatites.

same sample are substantial. Compared to pyrochlore from coarse-grained calcite-carbonatites they contain moderate to high contents of CaO (8.9–17.0 wt%), Na₂O (8.9–10.1 wt%), Ta₂O₅ (up to 2.5 wt%), ZrO₂ (up to 6.5 wt%), and ThO₂ (up to 2.0 wt%), variable amounts of REE oxides (2.1–18.9 wt%) and are low in UO₂ (b.d.l. – 1.6 wt%). Pyrochlore from medium- to fine-grained dolomite-carbonatites is relatively rich in CaO (11.5–18.7 wt%), but low in Na₂O (3.3–8.7 wt%) with similar contents of UO₂, ZrO₂, Ta₂O₅, ThO₂ and REE oxides as in medium- to fine-grained calcite-carbonatites.

Unzoned pyrochlores from the lapilli tuff (Henkenberg) show high values of CaO (15.9–20.0 wt%) and Na₂O (4.1–7.3 wt%). The contents of REE oxides (up to 5.2 wt%), ThO₂ (up to 0.8 wt%), UO₂ (up to 1.9 wt%), ZrO₂ (up to 2.7 wt%) and Ta₂O₅ (up to 0.9 wt%) are low.

The observed compositional variation in this large data set can be explained by the following set of substitution mechanisms that are of variable importance for the different textural types of pyrochlore (e.g., Kapustin, 1980; Hogarth, 1977; Hogarth et al., 2000; Electronic supplement ES11):



5.2.1. Depth-depending variations in the KB3 drill hole

The samples from surface outcrops combined with those from the KB3 drill cores allow for studying compositional variations in pyrochlore over a vertical depth of about 400 m within the Badberg body (Fig. 8). Whereas in the upper part of the sample profile, most element concentrations are roughly constant, compositional changes can be recognized below about 100 m above sea level (drill core meter 220–400). Below this depth, Na, Nb and F contents are relatively low, whereas U, REE, Ta, Fe, and Al contents are relatively high. Among the latter group, U contents gradually decrease in the lower part of the

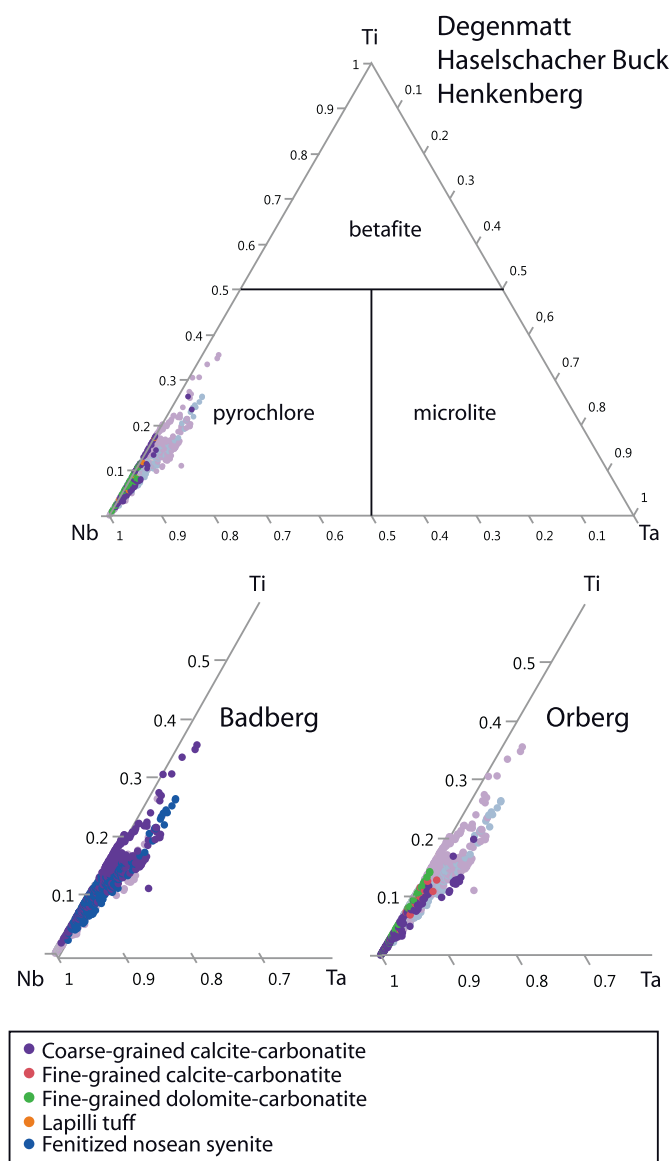


Fig. 6. Classification diagram for pyrochlore-group minerals with data from the various sample localities (after Hogarth, 1977). All compositions represent pyrochlore *senso stricto*.

profile, while REE contents increase further with depth. Other element contents either stay roughly constant throughout the profile (Ca, Sr, Mn, Y) or show rather unsystematic scatter (Zr, Ti, Th).

6. Discussion

6.1. Pyrochlore as a monitor for magmatic, hydrothermal and weathering processes

The textural and compositional variability in the Kaiserstuhl pyrochlores reveals important details on the magmatic and hydrothermal evolution of the various carbonatites of the complex. Similar textural complexity has been observed in several other occurrences of pyrochlore, as detailed below.

- (i) An early population of U-Ta-enriched pyrochlore, which is commonly resorbed and surrounded by oscillatory-zoned F-enriched pyrochlore has been interpreted to reflect a multi-stage magmatic evolution in several carbonatites worldwide (e.g., Chakhmouradian

and Zaitsev, 1999; Chakhmouradian and Mitchell, 2002; Zurevinski and Mitchell, 2004; Mitchell, 2005; Hogarth, 1977; Hogarth et al., 2000; Lee et al., 2006, Chakhmouradian, 2006). At the Kaiserstuhl, these U- and Ta-enriched pyrochlore cores are restricted to pyrochlores from some Badberg samples that occur in close proximity to fenitized nosean syenites. Therefore, we interpret the resorbed U- and Ta-rich cores as originally crystallized from a silicate magma and subsequently entrained in the carbonatite magma during emplacement. This is further supported by high Gamma Ray signatures related to U- and Ta-rich pyrochlore from the nosean syenites below the Badberg (Kirchheimer, 1973).

- (ii) Oscillatory-zoned F-enriched pyrochlores from all localities probably crystallized directly from the carbonatite magma. Oscillatory zoning in pyrochlore is seen as a typical magmatic feature, originating from higher crystal growth rates than diffusion rates for the respective elements in the melt (e.g., Hogarth et al., 2000).
- (iii) Pyrochlores from Orberg samples show intense patchy zonation and localized late-stage rims that crosscut earlier textures (Fig. 4F). These are exceptionally rich in Zr (up to 13.7 wt% ZrO₂), REE (up to 18.4 wt% REE oxides), Th (up to 6.5 wt% ThO₂), U (up to 8.2 wt% UO₂) and resemble pyrochlore and ceriopyrochlore compositions (Figs. 6 & 7; Hogarth, 1977). These textures are interpreted to reflect hydrothermal alteration. Such textures are much rarer in samples from the other localities, which may indicate a more evolved character of the Orberg rocks (see also below). Alternatively, this may be related to the much smaller size of the Orberg carbonatite bodies as it is presumably easier (in contrast to the much more competent sodalite monzogabbros in the vicinity of the Badberg) for Zr-REE-U-Th-rich fluids to ascent and percolate through the high-porosity polygenic diatreme breccias and into the carbonatite dykes and apophyses. Pyrochlores from the Kola peninsula evolve from initially Ca-Na-Th-rich towards Ba-Sr-(REE)-rich compositions by meteoric (metasomatic) fluids (Chakhmouradian and Williams, 2004) and a similar Ba-enrichment during hydrothermal stages has been described from the Sokli carbonatite in Finland (Lee et al., 2006). Late-stage REE-enriched pyrochlores and ceriopyrochlores from Kaiserstuhl are Ba-poor (always below detection limit). However, strong Ba-enrichment occurs in phlogopite from the Kaiserstuhl carbonatites (up to about 4.5 wt% BaO; Keller et al., 1990), while mica from the Sokli rocks (see above) is invariably Ba-poor (≤ 1.4 wt% BaO).
- (iv) Cracked and partly resolved pyrochlores typical of low-temperature (sub)surface weathering or late-stage hydrothermal alteration are characterized by A-site vacancies in the pyrochlore formula (e.g., Lumpkin and Ewing, 1995; Wall et al., 1996). At the Kaiserstuhl, pyrochlore data with significant A-site vacancies are depleted in Na-Ca-F. However, an increase in REE, Ti, Zr or Ba is not recognized, although this would be common for pyrochlores with high A-site vacancies related to a hydrothermal overprint (Lumpkin and Ewing, 1995). Hence, surficial weathering is seen as the dominant process that generates the A-site vacancies.

6.2. Fractionation of geochemical twins (Nb/Ta and Zr/Hf) during carbonatite evolution

Pyrochlores are likely to be the main carrier of Nb/Ta as supported by the rough correlation between the Nb/Ta ratios in pyrochlore and the respective whole-rock data (Fig. 9), especially considering the large variability of Nb/Ta ratios in the various textural types of pyrochlore (see above). Similarly, we assume that the major Zr and Hf hosts are pyrochlores as well, given the absence of notable Zr phases (such as baddeleyite or zircon) in the investigated samples. However, a direct comparison between whole-rock Zr/Hf ratios and pyrochlore data is not possible, as most EPMA analyses indicate Hf to be below the detection limit (~ 300 $\mu\text{g/g}$). Although some overlap exists, Orberg samples reach higher Nb/Ta and Zr/Hf ratios than samples from the Badberg and Nb

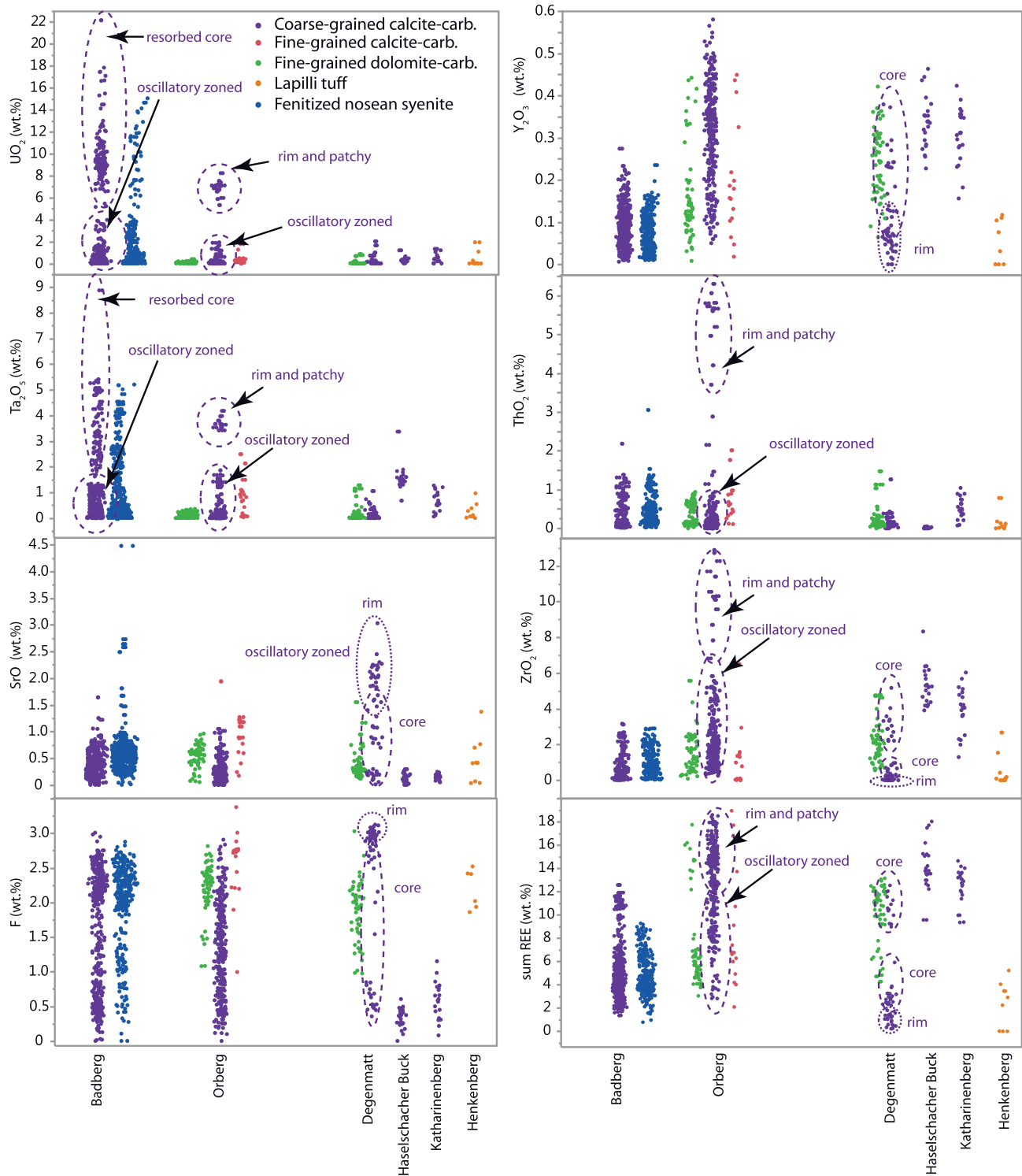


Fig. 7. Compositional variation of pyrochlore in the various Kaiserstuhl carbonatites and fenitized nosean syenites.

and Zr contents in Orberg samples are relatively high (Figs. 7 & 9), which is partly influenced by U- and Ta-enriched (and low Nb/Ta) resorbed pyrochlore cores in some Badberg samples and relatively high Zr contents in patchy-zoned REE-enriched pyrochlores and ceriopyrochlore from Orberg samples (Figs. 7 & 9). The significance of highly variable Nb/Ta and Zr/Hf ratios found in carbonatites (Fig. 5B & C) was discussed by Chakhmouradian (2006) and can be explained in the case of the Kaiserstuhl volcanic complex by fractional crystallization of pyrochlore. Starting with low Nb/Ta and Zr/Hf ratios and evolving

towards high ratios during differentiation would imply that the Orberg carbonatites represent more evolved rocks than those from Badberg if a common origin of these two bodies is assumed (Lehnert, 1989). Compared to silicate rocks, carbonatites of the Kaiserstuhl are enriched in Nb and Ta but depleted in Zr, and show a much larger spread, reaching very high Nb/Ta and Zr/Hf ratios (Fig. 5B & C). However, due to a lack of Nb/Ta and Zr/Hf ratios from the presumably carbonatite-related h a ynites and h a ynite melilitites (Schleicher et al., 1990), it is not possible to shed more light on the processes that generated carbonatites

Table 1
Representative pyrochlore analyses from the Kaiserstuhl.

Sample	HTAC_1232	HTAC_1253	HTAC_1353	HTAC_284	HTAC_1369	HTAC_1366	HTAC_1362
Texture	Oscillatory	Oscillatory	Oscillatory	Patchy	Unzoned	Patchy	Resorbed core
Location	Badberg	Orberg	Orberg	Has. Buck	Henkenberg	Degenmatt	Badberg
Na ₂ O	3.59	4.19	7.40	2.03	4.11	4.23	2.59
CaO	17.79	16.20	15.45	18.55	22.02	15.48	17.21
SrO	0.30	0.09	0.43	0.24	0.70	0.16	0.02
La ₂ O ₃	0.66	1.30	1.05	1.48	0.07	1.30	0.06
Ce ₂ O ₃	5.90	9.40	5.99	9.98	2.03	8.05	2.92
Pr ₂ O ₃	0.26	0.68	0.44	0.62	0.09	0.70	0.21
Nd ₂ O ₃	0.42	1.75	1.10	1.25	0.05	1.70	0.39
Sm ₂ O ₃	0.07	0.26	0.14	0.20	b.d.l.	0.29	0.09
Y ₂ O ₃	0.09	0.25	0.16	0.35	0.08	0.36	0.00
ThO ₂	b.d.l.	0.66	0.49	0.03	0.78	1.47	0.35
FeO	3.41	2.39	0.98	2.69	0.68	2.11	3.28
SiO ₂	b.d.l.	b.d.l.	0.00	b.d.l.	0.26	b.d.l.	0.02
MnO	0.16	0.39	0.14	0.33	0.07	0.63	0.02
ZrO ₂	0.08	4.60	2.62	3.91	0.42	4.77	0.29
TiO ₂	5.35	3.24	4.26	1.97	7.13	2.81	9.30
Nb ₂ O ₅	51.24	51.24	57.07	52.19	57.38	52.25	34.56
Ta ₂ O ₅	1.01	1.20	1.48	1.39	0.06	1.28	7.65
Al ₂ O ₃	0.16	0.13	0.03	0.20	b.d.l.	0.14	0.29
F	1.25	2.24	4.78	0.97	4.43	3.14	0.00
UO ₂	8.22	0.63	1.26	0.21	0.27	0.85	19.12
H ₂ O	1.80	1.22	0.22	1.73	0.48	0.76	2.29
F (corr.)	0.53	0.94	2.01	0.41	1.86	1.32	0.00
Total	99.95	100.85	7.40	98.58	100.63	101.72	98.37
Formulae on the basis of 2 B-site cations							
Na	0.45	0.53	0.91	0.26	0.50	0.53	0.36
Ca	1.24	1.13	1.05	1.32	1.47	1.07	1.32
Sr	0.01	0.00	0.02	0.01	0.03	0.01	0.00
La	0.02	0.03	0.02	0.04	0.00	0.03	0.00
Ce	0.14	0.22	0.14	0.24	0.05	0.19	0.08
Pr	0.01	0.02	0.01	0.02	0.00	0.02	0.01
Nd	0.01	0.04	0.02	0.03	0.00	0.04	0.01
Sm	0.00	0.01	0.00	0.00	0.00	0.01	0.00
Y	0.00	0.01	0.01	0.01	0.00	0.01	0.00
Th	0.00	0.01	0.01	0.00	0.01	0.02	0.01
Fe	0.19	0.13	0.05	0.15	0.04	0.11	0.20
Si	0.00	0.00	0.00	0.00	0.02	0.00	0.00
Mn	0.01	0.02	0.01	0.02	0.00	0.03	0.00
Zr	0.00	0.15	0.08	0.13	0.01	0.15	0.01
Ti	0.26	0.16	0.20	0.10	0.33	0.14	0.50
Nb	1.51	1.51	1.63	1.57	1.61	1.53	1.12
Ta	0.02	0.02	0.03	0.03	0.00	0.02	0.15
Al	0.01	0.01	0.00	0.02	0.00	0.01	0.02
F	0.26	0.46	0.95	0.20	0.87	0.64	0.00
U	0.11	0.01	0.02	0.00	0.00	0.01	0.27
REE	0.17	0.32	0.19	0.33	0.50	0.29	0.09
A-site	1.99	2.01	2.10	1.93	2.07	1.94	2.05
B-site	2.00	2.00	2.00	2.00	2.00	2.00	2.0
F (corr.) total	101.12	101.13	103.47	99.92	99.12	101.18	100.47

at the Kaiserstuhl volcanic complex (liquid immiscibility versus fractionation) in the context of partitioning of Nb, Ta, Zr and Hf between silicate and carbonatite melt (Veksler et al., 1998).

6.3. Pyrochlore compositions as an exploration indicator

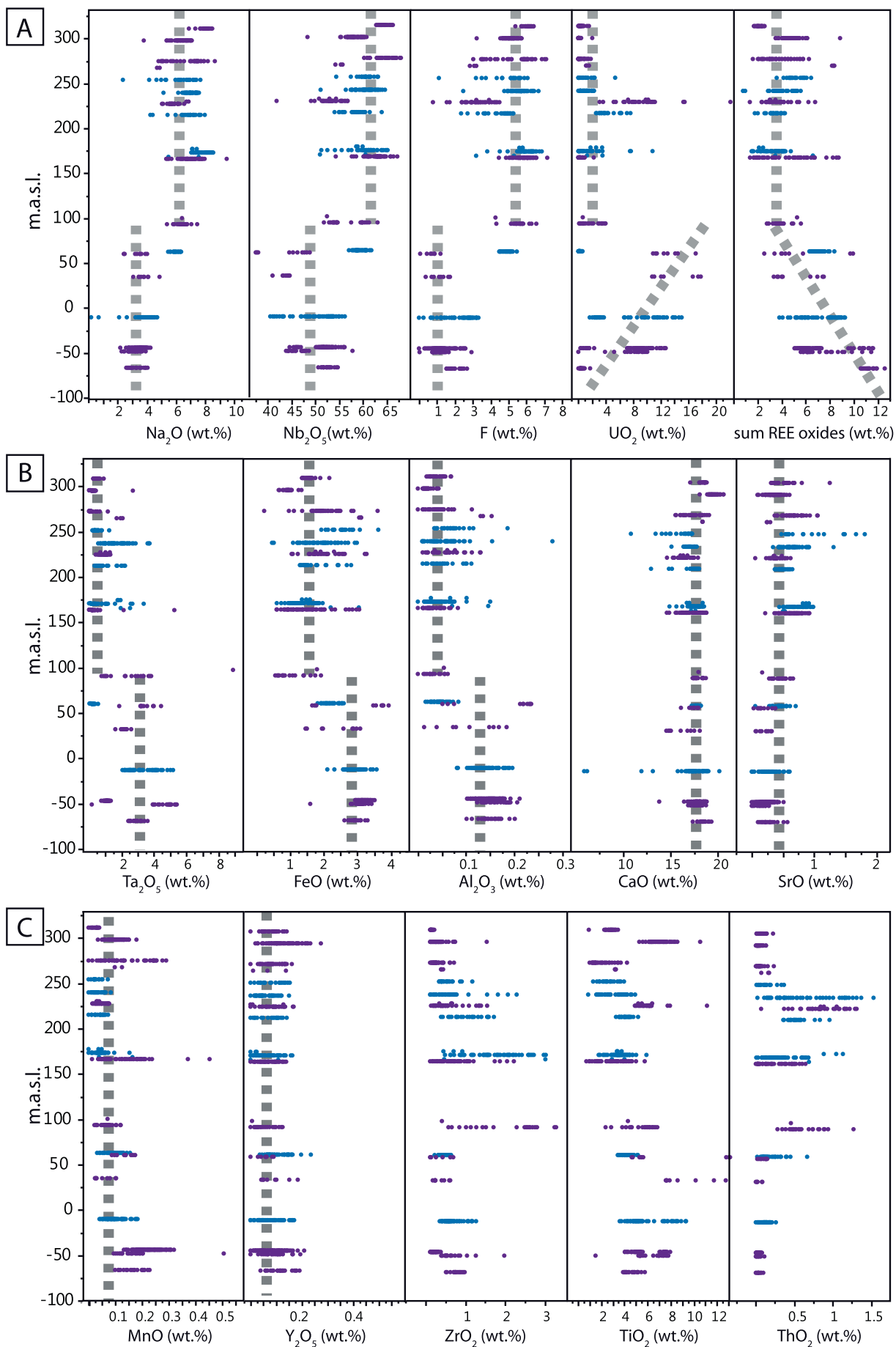
6.3.1. Depth-depending variations of pyrochlore composition

The depth-depending compositional variations in pyrochlore (see above) correlate well with geophysical data from the KB3 drilling that indicate increasing radioactivity at depth (Terra Tec, personal communication). The drill core meters below 100 m above sea level (drill core meter 220–400) reflects the central parts of a former vertical pipe structure (see above). In numerous elements (especially, Na, Nb, F, U,

TREES, Ta, Fe, Al) a significant and sharp compositional change in pyrochlore can be recognized. This may indicate a multi-stage emplacement of the carbonatitic magma or may reflect the combined effects of fractional crystallization and assimilation (AFC) as suggested by Giebel et al. (submitted). These results indicate that down-hole changes in pyrochlore compositions may have potential application as an exploration indicator.

6.3.2. Relation between intrusive and extrusive carbonatites

Extrusive carbonatites of the Kaiserstuhl contain fragments of fenitized nosean syenites and rare Nb-perovskite (see above). These mineralogical features only occur in some Badberg samples. We therefore suggest that the extrusive carbonatites at the western flank of the



(caption on next page)

Fig. 8. Compositional variation of pyrochlore down-hole the KB3 drilling. Note the variable behavior of the various elements below a depth of about 100 m above sea level.

complex may represent erupted equivalents of the hypothetical Badberg-Degenmat-Haselschacher pipe (Fig. 2B; see above) body. Therefore, extrusive carbonatites may offer the principal opportunity to predict HFSE mineralization expected in genetically related intrusive equivalents at depth. Hence, the pyrochlores from extrusive carbonatites of the complex indicate a minimum content of > 6 wt% total REE oxides (< 19 wt% in intrusive carbonatites) in pyrochlore in the ore body at depth, > 1 wt% ThO₂ (< 6.5 wt%), > 3 wt% ZrO₂ (< 14 wt%), > 1 wt% Ta₂O₅ (< 9 wt%), > 11 wt% TiO₂ (< 13.5 wt%), > 2 wt% UO₂ (< 22 wt%) and > 68 wt% Nb₂O₅ (< 69 wt%). However, the modal amounts of pyrochlores in the extrusive carbonatites are much lower than in the intrusive rocks. Other physical-chemical effects, such as hydrothermal alteration (which is markedly different in the Badberg and Orberg bodies, see above), differentiation stage of the carbonatitic magma during eruption/intrusion, will certainly affect such an estimation, especially if the geometry of the body at depth is insufficiently constrained. Nevertheless, this approach may be applied during first stages of exploration.

6.3.3. Element correlations of pyrochlore and whole-rock and their relation to geophysical exploration tools

In this section, the mineral chemical controls of geophysical exploration parameters will be discussed (Fig. 10). At Orberg, high REE (up to 2500 µg/g) and variable Th contents (up to 60 µg/g) in whole-rocks and in pyrochlores (up to 18.6 wt% and up to 6.3 wt%, respectively) do not strictly correlate (Fig. 10A). This is because apatite is an important REE carrier in the Kaiserstuhl carbonatites as well (Fig. 5D), with up to 7 wt% REE (Wang et al., 2014; Giebel et al., submitted), but typically low Th contents < 10 µg/g (Wang et al., 2014). Therefore, high Th contents may or may not indicate REE-enrichment in carbonatites and the common approach “if you look for REE, go for the Th” may not always be applicable. This result is problematic with respect to geophysical exploration as apatite is not sensitive to U-, Th-, K-mapping, magnetic susceptibility and geo-electrics and hence, apatite-hosted REE enrichment cannot be detected easily. However, a positive correlation between Th and Zr in whole-rocks (Fig. 10B) enables to detect enrichment of Zr by airborne Th-mapping, but may not help to detect Nb and Ta enrichment, due to their lack of correlation with Th (Fig. 10C & D).

In contrast to pyrochlore compositions (Fig. 7), no significant differences between U and Ta contents in whole rock samples from the Badberg (< 90 µg/g and < 40 µg/g) and Orberg (< 70 µg/g and < 60 µg/g) exist (Fig. 10E & F). This is presumably related to a much higher modal amount of pyrochlore at Orberg (with relatively low U and Ta contents) whereas at Badberg pyrochlore occurs in lower abundances, but with higher U and Ta contents (Fig. 7). However, the positive correlation between U and Ta in whole-rock and pyrochlore (Fig. 10E) indicates that Ta-rich samples could be mapped by e.g., airborne U-mapping. Hence, further studies have to be carried out to shed light on the mineralogical and mineral chemical factors which control geophysical signatures with a focus not only on Th (e.g., U mapping and U-Ta correlation in pyrochlore or apatite detection by geophysical parameters).

7. Summary and conclusion

The textural and compositional differences in pyrochlore-group minerals between intrusive and extrusive carbonatites and between different intrusive bodies allow for deciphering details of the magmatic, hydrothermal and meteoric evolution of the carbonatitic system of the Kaiserstuhl volcanic complex. In a more general sense, several results of this study can potentially be used as exploration indicators for carbonatite-related HFSE mineralization at depth:

- (i) The mineralogical composition of extrusive carbonatites of the Kaiserstuhl volcanic complex is similar to that of the coarse-grained calcite-carbonatites from Badberg, Degenmatt and Haselschacher Buck. Therefore, if extrusive carbonatites contain pyrochlore, a conservative estimate of HFSE content of pyrochlore in their intrusive equivalents appears possible.
- (ii) Pyrochlore from Badberg samples frequently contain U- and Ta-rich cores that probably represent xenocrystic material derived from underlying nosean syenites. Their subsequent crystallization history within the carbonatite magma itself is recorded by oscillatory-zoned F-rich pyrochlore that overgrew and partly resorbed these cores. Pyrochlore from Orberg samples, however, lacks such resorbed cores, suggesting the absence of such rocks below the Orberg area.

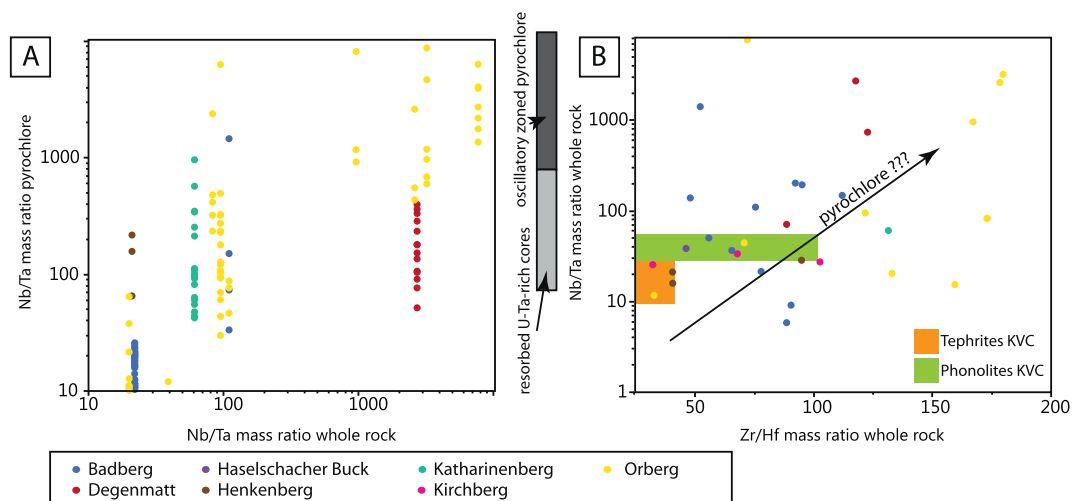


Fig. 9. Comparison of (A) Nb/Ta and (B) Zr/Hf ratios between pyrochlore and whole-rock data of Kaiserstuhl carbonatites.

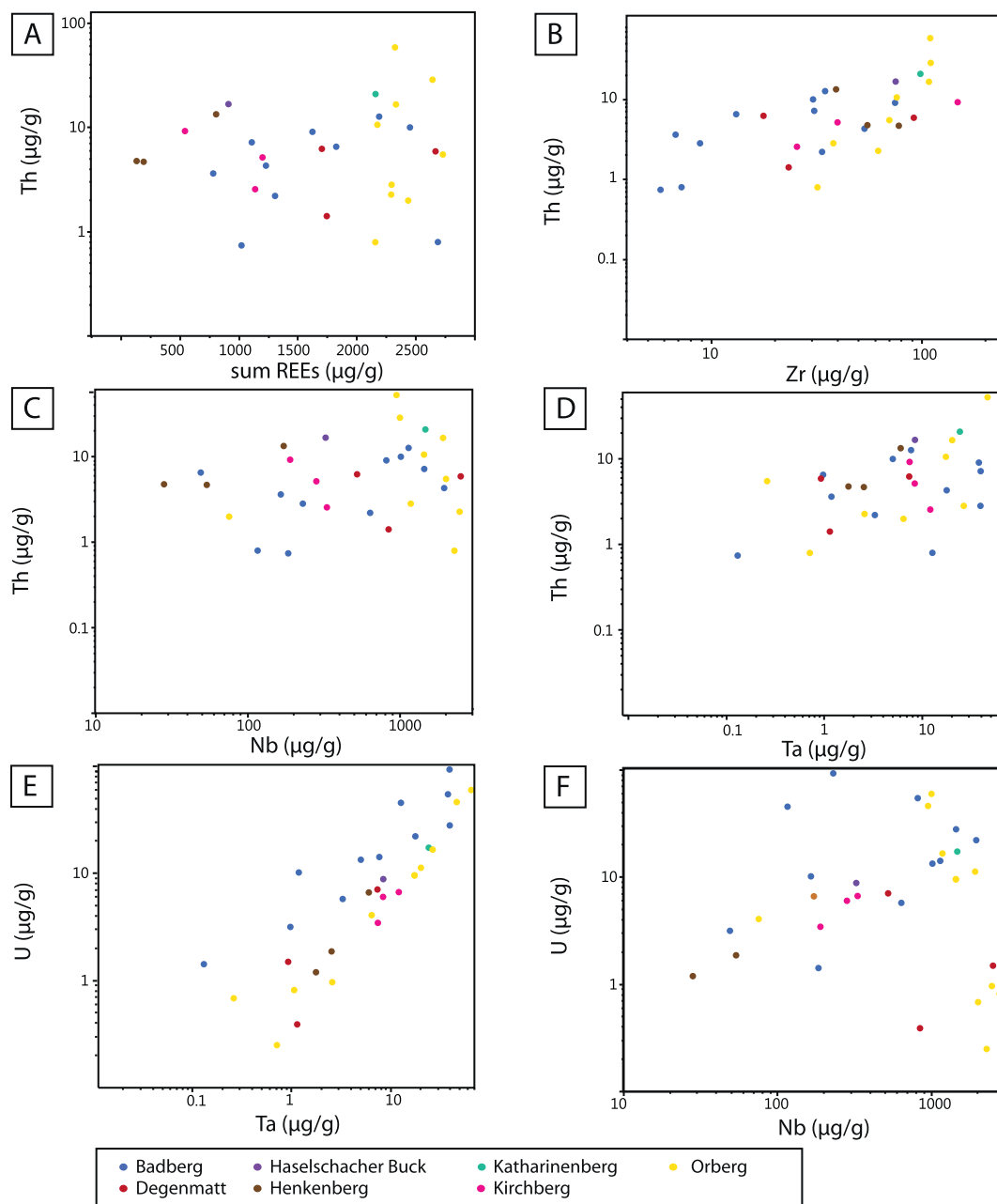


Fig. 10. (A–F) Correlations between the exploration parameters U and Th to Nb, Ta, Zr and REE. Note: In (B) and (E) Th is positively correlated to Zr and U to Ta.

(iii) The abundance of patchy zoned and Zr-REE-Th-U-enriched pyrochlore and ceriopyrochlore at the Orberg indicates a strong hydrothermal overprint of these rocks. They represent the highest values in all investigated samples, except for U, which is still higher in resorbed BSE-bright pyrochlore cores restricted to some of the Badberg samples.

Supplementary data to this article can be found online at <https://doi.org/10.1016/j.chemgeo.2018.08.008>.

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Appendix V

Accepted publication

Study 5

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Evidence for Magma–Wall Rock Interaction in Carbonatites from the Kaiserstuhl Volcanic Complex (Southwest Germany)

R. J. Giebel ^{1,2,*}, A. Parsapoor^{1,†}, B. F. Walter^{1,3}, S. Braunger¹, M. A. W. Marks¹, T. Wenzel¹ and G. Markl¹

¹Department of Geosciences, Eberhard Karls Universität Tübingen, Wilhelmstr. 56, 72074 Tübingen, Germany;

²Department of Geology, University of the Free State, 250 Nelson-Mandela-Drive, Bloemfontein 9300, South Africa;

³Institute for Applied Geosciences, Karlsruhe Institute of Technology, Adenauerring 20b, Karlsruhe 76131, Germany

*Corresponding author. Department of Geosciences, Eberhard Karls Universität Tübingen, Wilhelmstr. 56, 72074 Tübingen, Germany. Telephone: +49-(0)7071-29-73155. Fax: +49-(0)7071-29-3060. E-mail: r.j.giebel@gmx.de

[†]Joint first authorship.

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ABSTRACT

The mineralogy and mineral chemistry of the four major sövite bodies (Badberg, Degenmatt, Haselschacher Buck and Orberg), calcite foidolite/nosean syenite xenoliths (enclosed in the Badberg sövite only) and rare extrusive carbonatites of the Kaiserstuhl Volcanic Complex in Southern Germany provide evidence for contamination processes in the carbonatitic magma system of the Kaiserstuhl. Based on textures and composition, garnet and clinopyroxene in extrusive carbonatites represent xenocrysts entrained from the associated silicate rocks. In contrast, forsterite, monticellite and mica in sövites from Degenmatt, Haselschacher Buck and Orberg probably crystallized from the carbonatitic magma. Clinopyroxene and abundant mica crystallization in the Badberg sövite, however, was induced by the interaction between calcite foidolite xenoliths and the carbonatite melt. Apatite and micas in the various sövite bodies reveal clear compositional differences: apatite from Badberg is higher in REE, Si and Sr than apatite from the other sövite bodies. Mica from Badberg is biotite- and comparatively Fe²⁺-rich (Mg# = 72–88). Mica from the other sövites, however, is phlogopite (Mg# up to 97), as is typical of carbonatites in general. The typical enrichment of Ba due to the kinoshitalite substitution is observed in all sövites, although it is subordinate in the Badberg samples. Instead, Badberg biotites are strongly enriched in ^{IV}Al (eastonite substitution) which is less important in the other sövites. The compositional variations of apatite and mica within and between the different sövite bodies reflect the combined effects of fractional crystallization and carbonatite–wall rock interaction during emplacement. The latter process is especially important for the Badberg sövites, where metasomatic interaction released significant amounts of K, Fe, Ti, Al and Si from earlier crystallized nosean syenites. This resulted in a number of mineral reactions that transformed these rocks into calcite foidolites. Moreover, this triggered the crystallization of compositionally distinct mica and clinopyroxene crystals around the xenoliths and within the Badberg sövite itself. Thus, the presence and composition of clinopyroxene and mica in carbonatites may be useful indicators for contamination processes during their emplacement. Moreover, the local increase of silica activity during contamination enabled strong REE enrichment in apatite via a coupled substitution involving Si, which demonstrates the influence of contamination on REE mineralization in carbonatites.

Key words: contamination; carbonatite; Kaiserstuhl; mineral chemistry; calcite foidolite; REE mineralization; silicates in carbonatites

INTRODUCTION

Carbonatites are mantle-derived magmatic rocks with more than 30% primary igneous carbonate minerals (Mitchell, 2005; Woolley & Kjarsgaard, 2008). In most of the ~ 550 known occurrences they form intrusive bodies (e.g. plugs, pipes, dykes, sills), whereas extrusive equivalents, such as lavas and tuffs are notably scarce, with only about 50 known occurrences (Woolley & Church, 2005). Depending on their predominant carbonate mineral, carbonatites are subdivided into calcite- (sövites and alvikites), dolomite- (beforsites), ferro- and natro-carbonatites (Le Maitre *et al.*, 2002). Calcite is the most common magmatic carbonate (sometimes with exsolution of dolomite), magmatic dolomite and ankerite are less common and other carbonates (e.g. strontianite, barytocalcite) and REE-carbonates (e.g. burbankite, bastnäsite) are generally minor constituents (e.g. Zaitsev *et al.*, 1998; Moore *et al.*, 2015; Giebel *et al.*, 2017).

In addition to carbonate minerals, most carbonatites contain variable amounts of apatite, magnetite (spinel group minerals) and phlogopite and a number of additional silicate minerals (e.g. forsterite, monticellite, melilite, clinopyroxene, amphibole, garnet, feldspatoids, alkali feldspar) have been reported, although it may be difficult to distinguish primary carbonatitic crystals from xenocrysts (e.g. Andersen, 1988; Hogarth, 1989; Barker, 2001; Brod *et al.*, 2001; Chakrabarty *et al.*, 2009; Reguir *et al.*, 2012; Chakhmouradian *et al.*, 2017). In addition, various sulfides (pyrrhotite, pyrite, chalcopyrite, galena, sphalerite) and HFSE- and REE-rich minerals (e.g. baddeleyite/zircon, perovskite/titanite, pyrochlore, zirconolite, calzirtite) are typically present (e.g. Chakhmouradian, 2006; Farrell *et al.*, 2010; Gomide *et al.*, 2013; Bell *et al.*, 2015). Due to the high abundance of apatite and some of the above-mentioned REE- and HFSE-rich minerals, carbonatites are of general economic interest and several occurrences are currently mined (e.g. Araxa, Bayan Obo, Catalao, Kovdor, Mount Weld, Niobec; Gendron *et al.*, 1984; de Oliveira & Imbernon, 1998; Ivanyuk *et al.*, 2002; Krasnova *et al.*, 2004; Kanazawa & Kamitani, 2006; Yang *et al.*, 2011; Neumann & Medeiros, 2015).

Contamination by crustal or cogenetic intrusive rocks is generally not considered to play an important role during carbonatite magmatism, because carbonatitic melts have low densities and extremely low viscosities, enabling them to ascend rapidly (Treiman & Schedl, 1983; Jones *et al.*, 2013 and references therein). High-silica carbonatitic rocks in alkaline silicate-carbonatite complexes, which probably preserve the evidence of contamination of the parental magma, have often been ignored in previous studies, although there is evidence for the interaction between carbonatites and silicate rocks (Andrade *et al.*, 1999). However, potential contamination by silicate rocks in carbonatites cannot easily be detected by means of radiogenic isotope data (such as Sr, Nd and Pb isotope data) as

carbonatites often have high concentrations of these elements and their isotope systems are thereby 'buffered' against contamination with silicate rocks, which normally have much lower concentrations of these elements (e.g. Bell & Tilton, 2002 and references therein). The composition of silicate minerals in carbonatites and their comparison with country rocks, mantle rocks and associated ultramafic cumulates and alkaline rocks, however, may allow for their identification as true primary carbonatitic crystals or xenocrysts, but such studies are very rare (e.g. Vuorinen & Skelton, 2004).

Carbonatites from the Kaiserstuhl Volcanic Complex (South Germany) have been investigated previously (e.g. Keller, 1981; Hay & O'Neil, 1983; Hubberten *et al.*, 1988; Wimmenauer, 2003; Wang *et al.*, 2014; Teiber *et al.*, 2015; Braunger *et al.*, 2018; Walter *et al.*, 2018) but no systematic study of their mineralogy and mineral chemistry is available to date. We provide a comprehensive data set for the mineralogical inventory and the compositional variations of the major minerals in sövites and extrusive carbonatites, namely calcite, apatite, spinel group minerals (magnetite-magnesioferrite) and silicate minerals (mica, forsterite, monticellite, clinopyroxene, garnet). Based on the compositional variation of these phases, we tested the suitability of apatite and mica to assess the extent of carbonatite-wall rock interaction (contamination) in one of the sövite bodies of the Kaiserstuhl Volcanic Complex.

GEOLOGICAL SETTING AND OCCURENCE OF CARBONATITES IN THE KAISERSTUHL

The Kaiserstuhl Volcanic Complex (KVC) is situated in the Upper Rhine Graben (Fig. 1a), which is part of a larger extensional zone throughout central Europe that was established in a stress regime caused by Alpine orogenic processes in Tertiary times (e.g., Ziegler, 1982; Wilson & Downes, 1991). The emplacement of the KVC rocks along deep-reaching zones of weakness was enabled by lithospheric thinning (Hüttner, 1996; Edel *et al.*, 2006; Bourgeois *et al.*, 2007). The Upper Rhine Graben is characterized by numerous listric and mostly steep fault sets that are partly subparallel to the graben geometry. These faults and associated structures cause a horst and graben structure and a complex system of variably sized tectonic blocks (e.g. Beccaletto *et al.*, 2010).

The rocks of the KVC are of Miocene age (18–15 Ma; e.g. Kraml *et al.*, 2006 and references therein) and mainly consist of a tephritic to phonolitic rock series, accompanied by minor nephelinitic to limburgitic, and melilititic to haüynitic lithologies and carbonatites (e.g. Baranyi *et al.*, 1976; Keller *et al.*, 1990; Wimmenauer, 2003; Braunger *et al.*, 2018). The KVC rocks probably derive from several magma sources, with carbonatites being genetically related to the melilititic to haüynitic rocks (e.g. Schleicher *et al.*, 1990; Wang *et al.*, 2014).

The four major sövite bodies (Badberg, Degenmatt, Haselschacher Buck, Orberg) are spatially associated

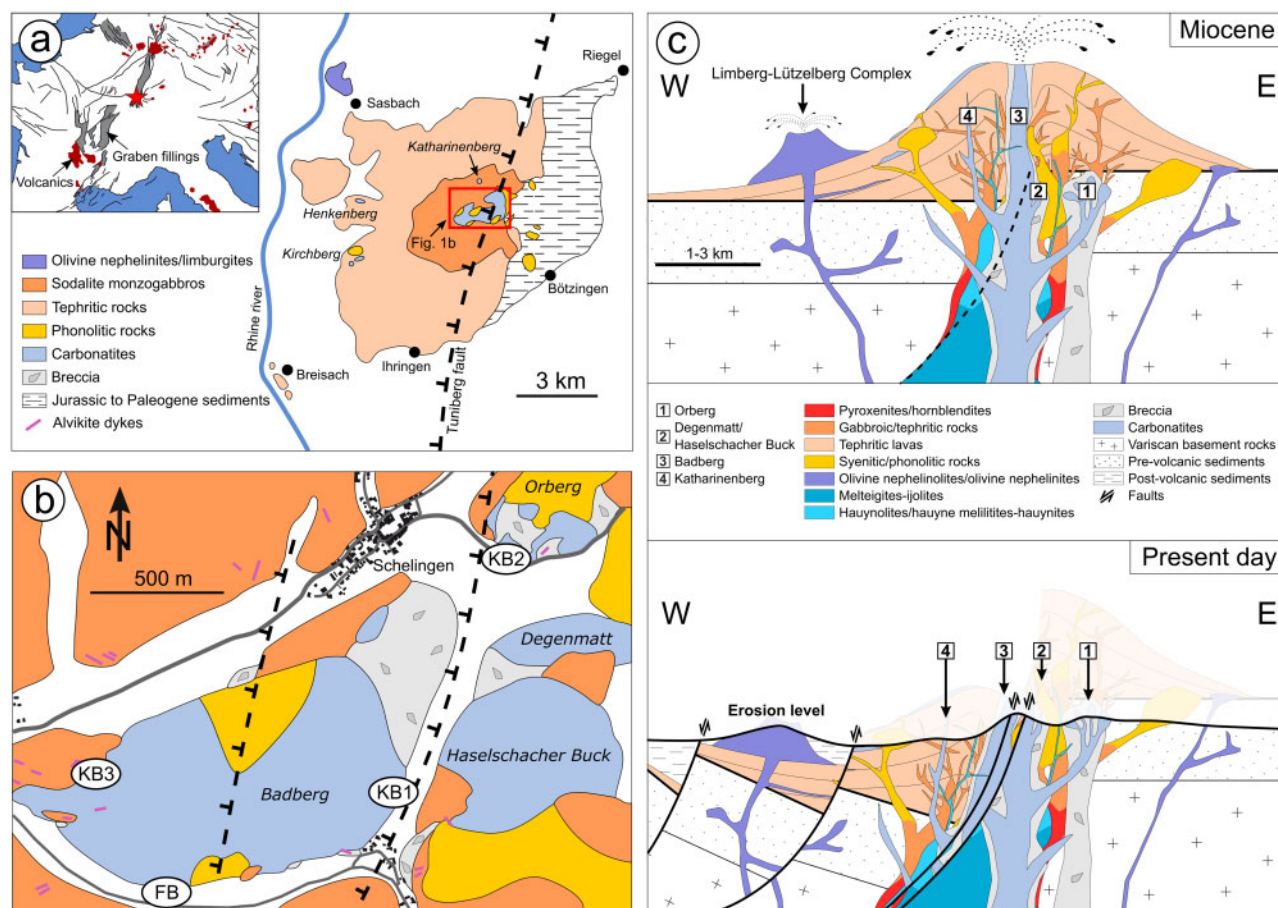


Fig. 1. Geology of the Kaiserstuhl Volcanic Complex (KVC). (a) Simplified geological map of the KVC (modified after Keller *et al.*, 1990) including an overview map of the European Cenozoic Volcanic Province (modified after Dèzes *et al.*, 2004), where the position of the KVC is indicated by a red star. (b) Detailed geological map of the central part of the KVC, which depicts several sövite bodies and the location of the four historic drill holes KB1, KB2, KB3 and FB. The schematic fault system is represented by a duplication of the Tuniberg normal fault. (c) Schematic cross section of the structural development of the carbonatites at the KVC.

with polygenetic breccias in the central area of the KVC and two small sövite occurrences (<10 m in diameter) are exposed at Katharinenberg and Kirchberg (Fig. 1a). The KVC is intersected by the regional Tuniberg Fault (Fig. 1), which causes a vertical displacement of ~1000 m north and ~3000 m south of the KVC, with a westwards down-throw towards the Rhine graben centre (Groschopf *et al.*, 1996; Beccaletto *et al.*, 2010). Geophysical evidence (seismic surveys; e.g. Brauch *et al.*, 2018 and references therein) suggests at least one sub-parallel structure further W that intersects the Badberg sövite body (Fig. 1b), but the vertical displacement of this structure is unknown.

The about 400 m thick, pipe-like Badberg body is banded and inclined at about 60° towards the northwest. It intrudes sodalite monzogabbros and phonolites and is partly underlain by foid syenites and phonolites as indicated by the scientific KB3 and FB drillings (Kirchheimer, 1973; Blust, 1993). Mineral banding in the sövite is subparallel to its contacts towards the country rocks and numerous cm- to m-sized rafts of calcite foidolites occur aligned parallel to the banding. The relative abundance of these xenoliths decreases from the

top of the tilted Badberg body towards its deeper parts. The sövite bodies at Degenmatt and Haselschacher Buck probably reflect similar pipe-like geometries but are not exposed well enough to constrain this. However, geophysical data (gravimetric and magnetic) indicate a continuous pipe-like body below the Degenmatt/Haselschacher Buck bodies, whereas there is no evidence that the Badberg body continues to greater depth (Brauch *et al.*, 2018).

Sövites at Orberg are exposed in five quarries (designated as I–V) and display variable geometries. Sövite sills dipping by 25° towards the northeast are exposed in quarries I and II, quarry III exposes a 10 m thick sheet-like body with a dip of 70° towards the northwest, and quarries IV and V show irregular geometries implying a cauliflower-like structure. The emplacement of the Orberg sövites was probably promoted by distinct zones of weakness, reflected by pre-existing polygenic breccias (proved by KB2 scientific drilling in 1991 in quarry II; Hubaux, 1964).

Dolomite-bearing alvikites (still calcite-dominated, though) occur as fine- to medium-grained subvertical dykes, crosscutting all rock types and rare dolomite-

dominated varieties (beforsites) are known (e.g. Van Wambeke, 1964; Sigmund, 1996). Alvikites show variable thicknesses (cm- to m scale) and textures. They have been described in great detail previously (e.g. Katz & Keller, 1981; Sommerauer & Katz-Lehnert, 1985; Katz-Lehnert, 1989) and are not subject to the present study.

At Henkenberg and Kirchberg (Fig. 1a), three layers (each 1 to 1.5 m thick) of extrusive carbonatite, represented as alternating sequences of carbonatitic crystal tuff, lapillistones and lavas and interbedded with tephritic and basanitic pyroclastics, are preserved (Keller, 1978; Keller, 1981, Keller, 1989; Woolley & Kjarsgaard, 2008). The distance between these two localities is about 1.5 km, but the original extent of these deposits is unknown, due to intense anthropogenic land transformations in the 1970s.

SAMPLE MATERIAL AND ANALYTICAL METHODS

The material investigated for this study is part of the sample set ($N \approx 400$) of Walter *et al.* (2018) and includes 46 carefully selected samples from surface outcrops and from the scientific KB2 and KB3 drillings, covering the major sövite bodies and the two known occurrences of extrusive carbonatites in the KVC. A complete sample list is given in Table 1, along with locations, GPS coordinates and petrographic details.

Mineral compositions were acquired using a JEOL 8900 electron microprobe at the Institute of Geosciences, Universität Tübingen, Germany. Acceleration voltage was 15 kV (for apatite, phlogopite, clinopyroxene and carbonate) and 20 kV (for garnet and spinel group minerals) using a beam current of 10 nA (for apatite) and 20 nA (for other minerals). Depending on the mineral, the beam diameters varied from a focused beam (for olivine, spinel group minerals, pyroxene and zirconolite), 2 μm (mica), 5 μm (carbonates and garnet) to 10 μm (apatite). Natural and synthetic standards were used for calibration (see ES1, Supplementary Data; supplementary data are available for downloading at <http://www.petrology.oxfordjournals.org>). Data reduction was performed using the internal ZAF and $\phi\rho z$ matrix correction software of JEOL (Armstrong, 1991). Details on the WDS configuration used and typical detection limits for the individual elements of various minerals are given in the Supplementary Data; supplementary data are available for downloading at <http://www.petrology.oxfordjournals.org> (ES1).

PETROGRAPHY

Sövites

Besides calcite, the most abundant minerals in the different sövites are apatite, mica, spinel group minerals and pyrochlore, with highly variable modal amounts (Table 1; Fig. 2) on a dm- to m-scale. Occasionally, mineral banding represented by alternating coarse-grained and finer grained layers of non-carbonate minerals

occurs in some areas. Rarely, thin olivine-rich layers (Orberg) and in one case, a perovskite-rich layer (Badberg, see below) have been exposed. However, the major mineralogical differences between the sövite bodies are reflected by the presence/absence of clinopyroxene, olivine and monticellite.

Olivine is present in two samples from Haselschacher Buck and Katharinenberg and occurs as up to 500 μm large euhedral to subhedral grains, which are often serpentinized along cracks and grain boundaries and are occasionally replaced by calcite (Fig. 3a). Subhedral to anhedral monticellite occurs in one sample from Orberg V (Fig. 3b).

Apatite is the most abundant non-carbonate mineral and occurs in all investigated sövites. Textures range from large radial crystal aggregates (Fig. 3c), medium-sized prismatic grains to anhedral patches mostly enclosed by calcite and less commonly by phlogopite and spinel group minerals. Most apatites show variable zoning patterns, ranging from oscillatory zonation (Fig. 3d; with occasional resorbed cores) to patchy zonation.

Clinopyroxene was only observed in some of the Badberg samples, where it is mostly euhedral but typically cracked and altered along cleavages. In some cases it shows rounded grain boundaries towards calcite (Fig. 3e).

The abundance and general appearance of mica (Figs 3e and 4) differs between samples from Badberg (commonly very mica-rich) and the other localities (generally mica-poor). With the exception of the Badberg samples, mica occurs as <2 mm large euhedral to subhedral laths to stubby grains, mostly enclosed by calcite (Fig. 4a). It is almost colourless under plane-polarized light and is occasionally altered to chlorite. Mica from Orberg III and Katharinenberg is typically oscillatory zoned (Fig. 4b), while mica from the Degenmatt samples is less so, and mica from other localities lacks any visible zonation. Mica from the Badberg samples is much larger (up to 2 cm), exhibiting an olive-green to khaki appearance under plane-polarized light. Two different modes of occurrence in the Badberg samples are distinguished: (1) blocky crystals accumulated at the contact towards the calcite foidolites (Fig. 4c; see below) are classified as 'black-wall mica', whereas (2) bundles of large mica laths occur dispersed in the Badberg samples and are seemingly independent of the presence of calcite foidolite relicts (Fig. 4d).

Spinel group minerals are very common and generally occur as several mm-sized disseminated subhedral to euhedral grains (occasionally very porous; Fig. 3f). They are mostly enclosed by calcite and occasionally contain inclusions of apatite. In very few samples, however, they are overgrown by apatite. Abundant pyrochlore (see details in Walter *et al.*, 2018) and rare zirconolite (Van Wambeke, 1964; Sinclair & Eggleton, 1982; Keller, 1984) crystallized more or less simultaneously with spinel group minerals. Zirconolite was only found in a sample from Haselschacher Buck, where it

Table 1: List of samples investigated in this study along with their major mineralogy

Locality	Rock type	Sample #	Depth	UTM coordinates	Ap	Phl	Mgt	Pcl	Cpx	Grt	Ol/ Mtc	Cal	Dol
Badberg	Calcite foidolite	HTAC 260	Surface	32 U 401310 5327748	x	x	-	(tr)	X*	X	-	x	-
Badberg	Sövite	HTAC 1136	KB3 (267 m)	32 U 400940 5327843	X	X	X*	-	X*	-	-	X	-
Badberg	Calcite foidolite	HTAC 1139	KB3 (254 m)	32 U 400940 5327843	x*	x	-	-	X	X*	-	x	-
Badberg	Calcite foidolite	HTAC 1147	KB3 (241 m)	33 U 400940 5327844	x*	x	-	-	X	X*	-	x	-
Badberg	Calcite foidolite	HTAC 1148	KB3 (240 m)	32 U 400940 5327843	x	x*	-	(tr)	X	X	-	x	-
Badberg	Sövite	HTAC 1151	KB3 (227m)	32 U 400940 5327843	X	X	x	x	x*	-	-	X*	-
Badberg	Sövite	HTAC 1156	KB3 (215 m)	32 U 400940 5327843	X	X*	x	x	-	-	-	X	-
Badberg	Calcite foidolite	HTAC 1158	KB3 (204 m)	32 U 400940 5327843	x	X	-	(tr)	X	X*	-	X	-
Badberg	Calcite foidolite	HTAC 1168	KB3 (177 m)	32 U 400940 5327843	x	x	-	(tr)	X	X*	-	x	-
Badberg	Calcite foidolite	HTAC 1173	KB3 (165 m)	32 U 400940 5327843	x	X	x	-	X	X*	-	x	-
Badberg	Calcite foidolite	HTAC 1178	KB3 (150 m)	32 U 400940 5327843	x*	-	-	(tr)	X*	X	-	X	-
Badberg	Calcite foidolite	HTAC 1197	KB3 (100 m)	32 U 400940 5327843	x	x	-	(tr)	X*	X	-	x	-
Badberg	Sövite	HTAC 1200	KB3 (93m)	32 U 400940 5327843	X	X*	x	(tr)	-	-	-	X	-
Badberg	Calcite foidolite	HTAC 1200	KB3 (93 m)	32 U 400940 5327843	x	X*	x	-	X	X*	-	x	-
Badberg	Sövite	HTAC 1215	KB3 (35 m)	32 U 400940 5327843	X	x	X	x	x*	-	-	X	-
Badberg	Sövite	HTAC 1229	KB3 (-10 m)	32 U 400940 5327843	X*	X	X	x	X*	-	-	X*	-
Badberg	Sövite	HTAC 1237	KB3 (-48 m)	32 U 400940 5327843	X	X*	X	(tr)	x*	-	-	X*	-
Badberg	Sövite	HTAC 1245	KB3 (-66 m)	32 U 400940 5327843	x	X*	X	(tr)	x	-	-	X	-
Badberg	Sövite	HTAC 1362	Surface	32 U 400763 5327907	X	x	X	(tr)	x	-	-	X*	-
Orberg II	Sövite	HTAC 1253	KB2 (4 m)	32 U 402477 5328680	x*	x	X	x	-	-	-	X*	x
Orberg II	Sövite	HTAC 1267	KB2 (271 m)	32 U 402477 5328680	x	x	x	-	-	-	-	X*	x
Orberg II	Sövite	HTAC 1318	KB2 (191.7 m)	32 U 402477 5328680	x	x*	x	-	-	-	-	X	x
Orberg II	Sövite	HTAC 1282	KB2 (242 m)	32 U 402477 5328680	x	x	x	-	-	-	-	X*	x
Orberg II	Sövite	HTAC 1294	KB2 (323 m)	32 U 402477 5328680	x	x	X	x	-	-	-	X*	-
Orberg III	Sövite	HTAC 1410	Surface	32 U 402555 5328668	x*	x*	x*	x	-	-	-	X*	-
Orberg III	Sövite	HTAC 1413	Surface	32 U 402555 5328668	x*	x*	x*	x	-	-	-	X	-
Orberg IV	Sövite	HTAC 0237	Surface	32 U 402691 5328724	X*	x*	X*	x	-	-	-	X*	-
Orberg IV	Sövite	HTAC 0239	Surface	32 U 402691 5328724	X*	-	x	x	-	-	-	X*	-
Orberg IV	Sövite	HTAC-1353	Surface	33 U 402691 5328724	x*	x	x*	x	-	-	-	X	-
Orberg V	Sövite	HTAC 0222	Surface	32 U 402640 5328781	x*	x*	-	x	-	-	-	X	X
Orberg V	Sövite	HTAC 0224	Surface	32 U 402640 5328781	X	-	X*	X	-	-	-	X	-
Orberg V	Sövite	HTAC 0225	Surface	32 U 402640 5328781	x*	x*	X*	X	-	-	-	X*	-
Orberg V	Sövite	HTAC 1356	Surface	32 U 402640 5328781	x*	-	x	-	-	-	-	X*	-
Orberg V	Sövite	HTAC 1366	Surface	32 U 402620 5328185	x*	x*	-	x	-	-	-	X*	-
Degenmatt	Sövite	HTAC 1399	Surface	32 U 403779 5326957	X*	x*	-	x	-	-	-	X*	-
Haselschacher	Sövite	HTAC 0284	Surface	32 U 402668 5327797	X*	x*	x*	x	-	-	-	X*	-
Buck													
Haselschacher	Sövite	HTAC 1354	Surface	32 U 402668 5327797	x*	x*	x*	-	-	-	x*/-	X*	-
Buck													
Katharinenberg	Sövite	HTAC 1415b	Surface	32 U 402288 5329644	X	x*	X*	x	-	-	-	X*	-
Katharinenberg	Sövite	HTAC 1415c	Surface	32 U 402288 5329644	X*	x*	x*	x	-	-	X*/-	X*	-
Henkenberg	Crystal tuff	GM1	Surface	32 U 396571 5327645	x	-	-	-	x*	x*	-	X*	-
Henkenberg	Crystal tuff	GM2	Surface	32 U 396571 5327645	x	-	x*	(tr)	X*	x*	-	X*	-
Henkenberg	Crystal tuff	GM3	Surface	32 U 396570 5327645	x*	x	x*	-	X	x*	-	X	-
Henkenberg	Crystal tuff	HTAC 1369	Surface	32 U 396571 5327640	x	-	x	-	-	x*	-	X	-
Henkenberg	Crystal tuff	HTAC 1371	Surface	32 U 396571 5327640	x	-	x*	-	-	x	-	X	-
Henkenberg	Lapilli stone	GM4	Surface	32 U 396570 5327647	x	-	x*	-	x*	x*	-	X*	-
Kirchberg	Lava	HTAC 1337	Surface	32 U 396571 5327640	-	-	x*	-	-	x*	-	X*	-

X, major component; x, minor component; (tr), trace component.

*, EPMA data available.

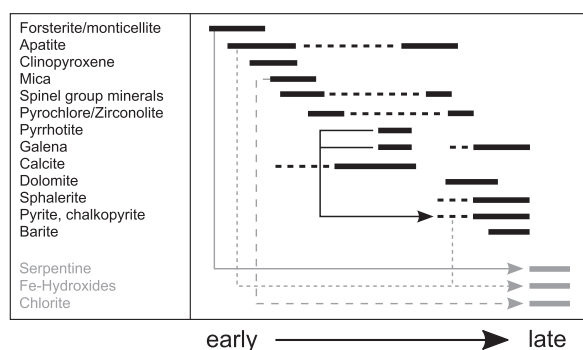


Fig. 2. Paragenetic scheme for sövites of the KVC including major alteration phases (in grey).

occurs as up to 30 µm large and occasionally zoned crystals (Fig. 4g). Perovskite (in the old literature called 'dysanolyte') was reported from a thin horizon at the Badloch quarry (southern Badberg; [Wimmenauer, 1963](#); [Hornig-Kjarsgaard, 1998](#)). This outcrop is, however, lost and none of the investigated samples contains perovskite.

The main phase of calcite formation occurred relatively late, only rarely calcite occurs as small inclusions in apatite. Calcite exhibits equigranular and subhedral textures, with grain sizes varying from 100 µm up to several cm. Late-stage dolomitization of calcite is occasionally observed (Fig. 4h).

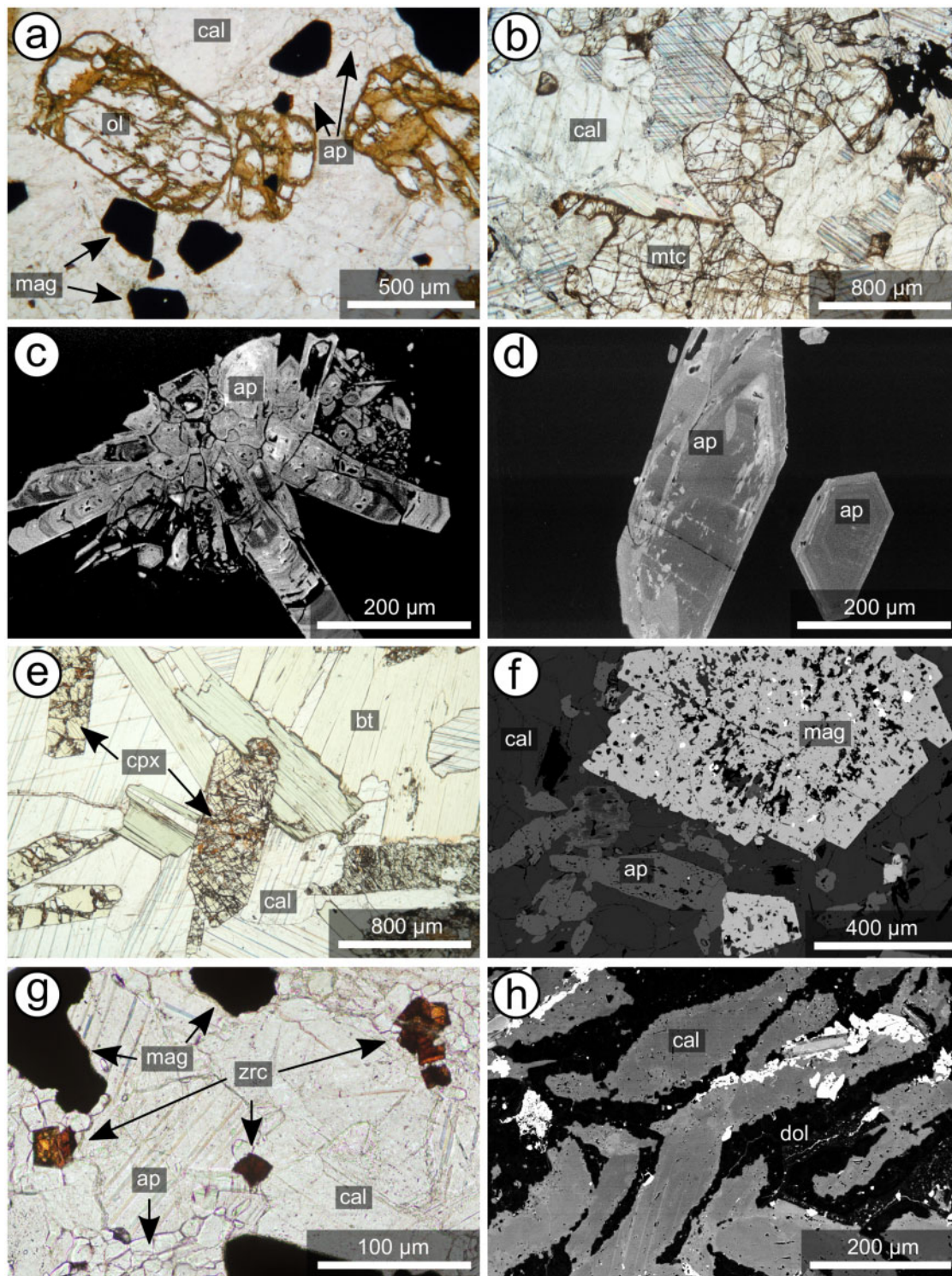


Fig 3. Photomicrographs (plane polarized light) and backscattered electron (BSE) images of the mineral assemblage of sövites from the KVC. (a) Coarse-grained olivine altered to serpentine (Haselschacher Buck). (b) Subhedral to anhedral monticellite (Orberg V). (c) Stellar aggregates of apatite (Orberg V). (d) REE- and Si-rich rims with REE- and Si-poor cores in apatite (Badberg). (e) Partly resorbed and altered clinopyroxene associated with biotite (Badberg). (f) Porous spinel group mineral (magnetite–magnesioferrite) associated with apatite and calcite (Orberg III). (g) Zirconolite (zrc) in olivine-bearing sövite (Haselschacher Buck). (h) Subhedral calcite surrounded and partly replaced by late-stage dolomite (Orberg II).

Abbreviations: ap, apatite; bt, biotite; cal, calcite; cpx, clinopyroxene; dol, dolomite; mag, spinel group mineral (magnetite–magnesioferrite); mtc, monticellite; ol, olivine; zrc, zircon.

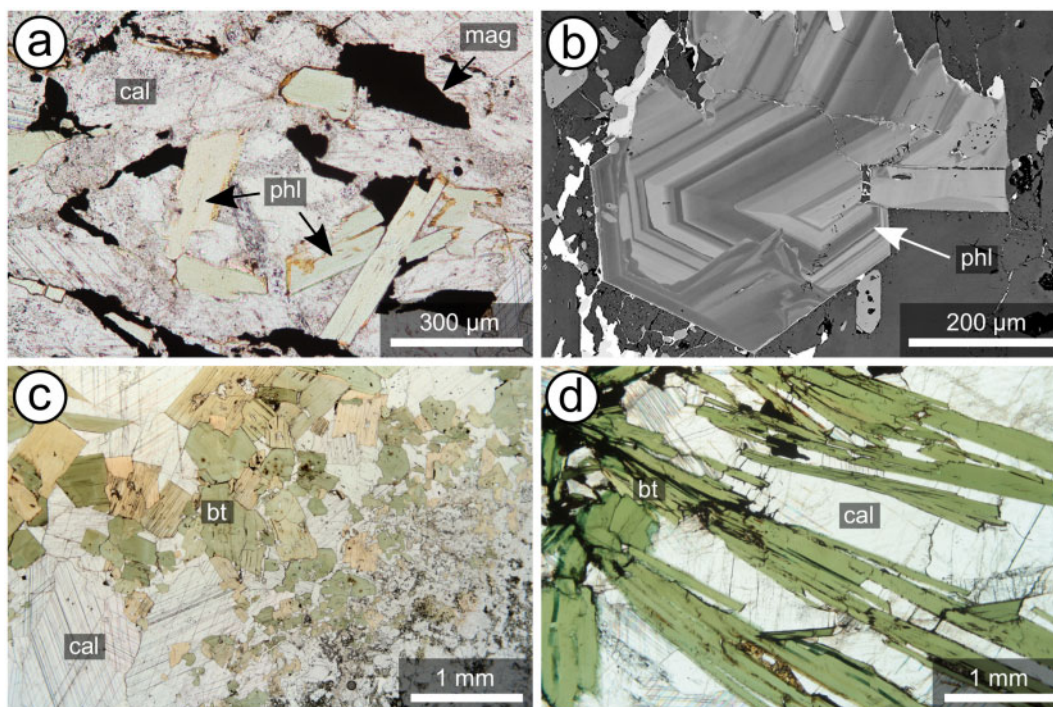


Fig. 4. Photomicrographs (plane polarized light) and backscattered electron (BSE) images of the different appearances of mica in sövites from the KVC. (a) Euhedral to subhedral laths of phlogopite enclosed by calcite and spinel group minerals (magnetite-magnesioferrite; Orberg V). (b) Oscillatory zoned Ba-rich phlogopite (Orberg III). (c) Coarse-grained 'black-wall' biotite (Badberg). (d) Coarse-grained bundles of biotite laths independent of the presence of calcite foidolite (Badberg). Abbreviations (see Fig. 3): bt, biotite; phl, phlogopite.

Rare pyrrhotite crystallized before calcite, but other sulphides (mainly pyrite, with minor galena and sphalerite) are of late-stage/hydrothermal origin and are often associated with fine-grained and interstitial barite.

Extrusive carbonatites

Carbonatite lavas have only been observed at the Kirchberg (Fig. 1). They consist of up to 0.6 mm large oscillatory zoned calcite laths and up to 150 µm large euhedral spinel group mineral grains, set in a fine-grained groundmass of calcite, apatite and spinel group minerals (Fig. 5a). In some cases, the spinel group minerals show thin rims, which are BSE-dark (Fig. 5b). Occasional xenocrysts of nepheline (300 µm; Fig. 5a) and garnet (200 µm) have been observed.

Carbonatite lapillistones and crystal tuffs from the Henkenberg were described in detail by Keller (1981, 1989). Lapillistones consist of calcite laths and sub-rounded lapilli cemented by fine-grained secondary calcite. The 0.5 to 10 mm large lapilli contain, in addition to calcite, larger amounts of spinel group minerals and prismatic apatite, and rare grains of clinopyroxene and garnet (Fig. 5c). Crystal tuffs are mainly composed of fragments of calcite (showing mosaic textures), spinel group minerals, garnet and pyroxene, with rare apatite and pyrochlore, set in a groundmass of calcite (Fig. 5d). Occasionally intercalated clasts of calcite foidolite and sövite can be recognized.

Calcite foidolites

Xenoliths of calcite foidolite (cm- to m-sized) occur only in the Badberg sövite (and rarely in carbonatitic tuff layers of the Henkenberg). Their abundance decreases towards the centre of the Badberg body, where they are strongly disaggregated compared to those in marginal zones. Sövite that contains strongly disaggregated xenoliths is exceptionally mica-rich (Fig. 6). Based on drill core loggings (Blust, 1993), the average carbonatite/xenolith ratio is about 5:1, but varies strongly from about 2:1 (marginal areas) to about 50:1 (central areas) over a range of tens of metres.

The contact between calcite foidolites and the surrounding sövite is characterized by coarse-grained seams of olive-green to khaki mica (see above) and an occasional transition zone consisting of strongly altered garnet and clinopyroxene, frequently intruded by carbonatitic veins (Fig. 7a). The central parts of these xenoliths consist of variable amounts of altered foid minerals (see below) and relict feldspar, with variable amounts of clinopyroxene, garnet and interstitial calcite (Fig. 7b–d).

Former nosean or häuyné (now decomposed to various zeolite minerals and calcite) is the most common phase. It mostly forms rounded grains of variable sizes and only rarely, euhedral grains containing small Fe-sulfide/oxide micro-inclusions are present (Fig. 7c). Rare relics of alkali feldspar are variably replaced by zeolites, calcite, mica and other minerals (Fig. 7d).

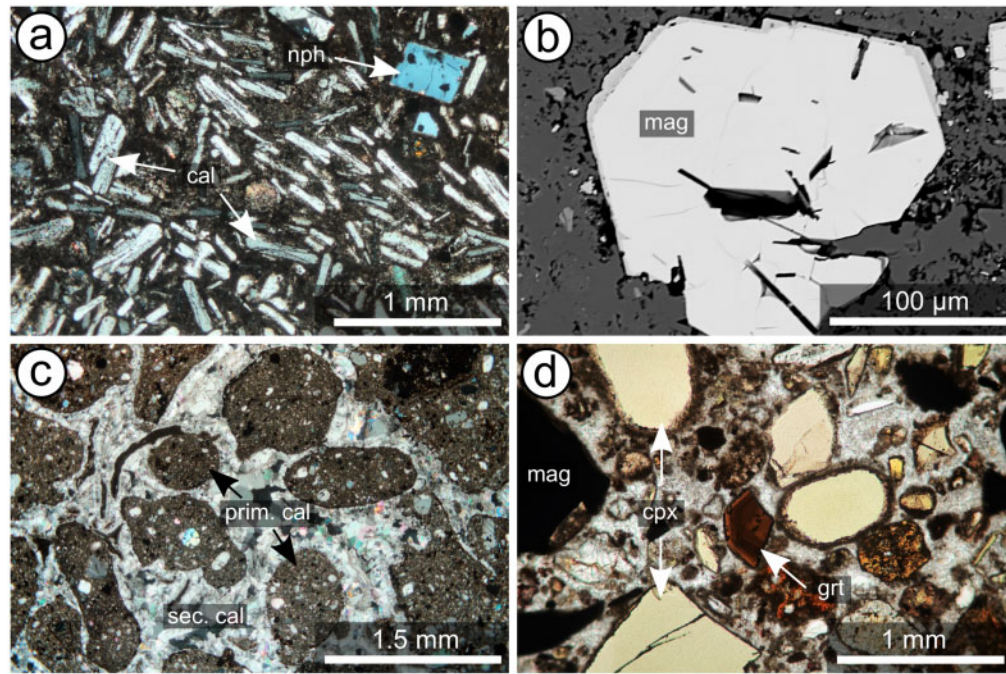


Fig. 5. Photomicrographs (plane and crossed polarized light) and backscattered electron (BSE) images of textural features of extrusive carbonatites from the KVC. (a) Carbonatite lava with calcite laths in a matrix of fine-grained calcite, apatite and spinel group minerals (magnetite–magnesioferrite). (b) Euhedral spinel group minerals with darker rims (enriched in Al) in carbonatite lava. (c) Lapillistone porphyritic lapilli in a fine-grained groundmass of calcite, spinel group minerals and apatite, cemented by secondary calcite. (d) Crystal tuff with fragments of clinopyroxene, garnet (grt) and spinel group minerals in groundmass of calcite.

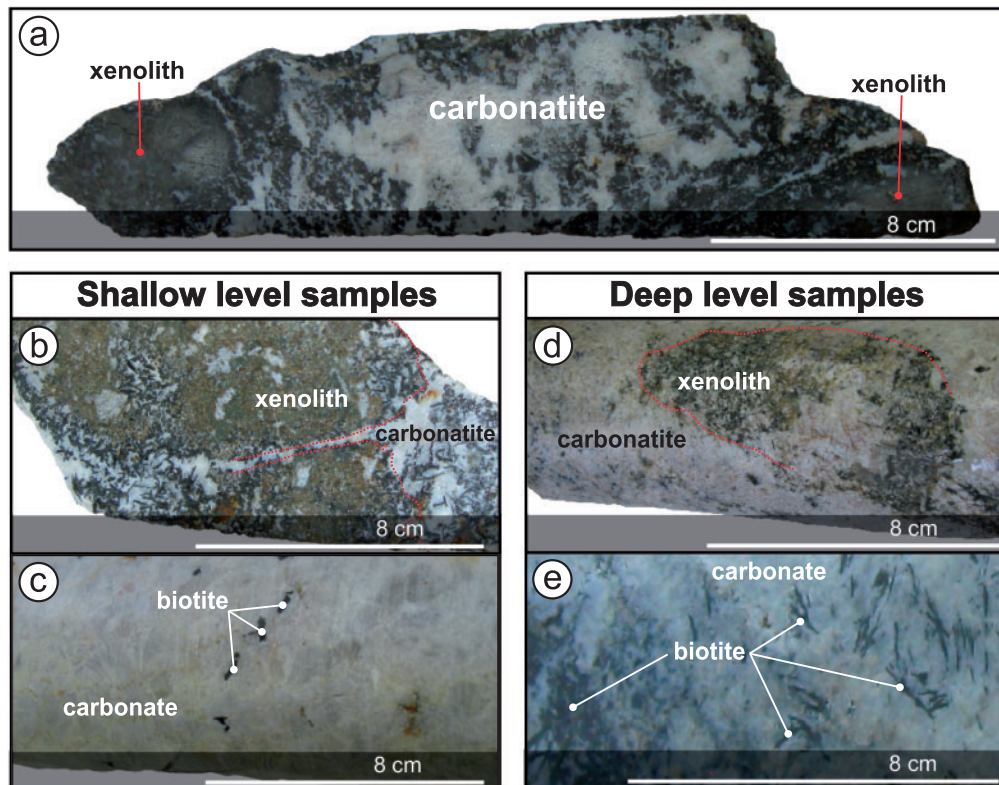


Fig. 6. Xenoliths in the Badberg carbonatites. (a) Carbonatite with a high xenolith proportion and strong proximal (to the xenolith) mica formation. (b) Xenolith from a 'shallow' drill core level (reflecting the marginal zone of the carbonatite pipe) showing significant mica formation around (black-wall mica) and in direct vicinity of the xenoliths. (c) Sections distal to a xenolith within a 'shallow' drill core level showing low mica formation. (d) Xenolith from a 'deep' drill core level (reflecting the central zone of the carbonatite pipe) showing strong disaggregation. The xenolith is nearly almost replaced by calcite. (e) Sections distal to xenolith within a 'deep' drill core level showing significant mica formation. The correlation between stronger disaggregated/resorbed xenoliths and the increased formation of mica shows a clear dependency.

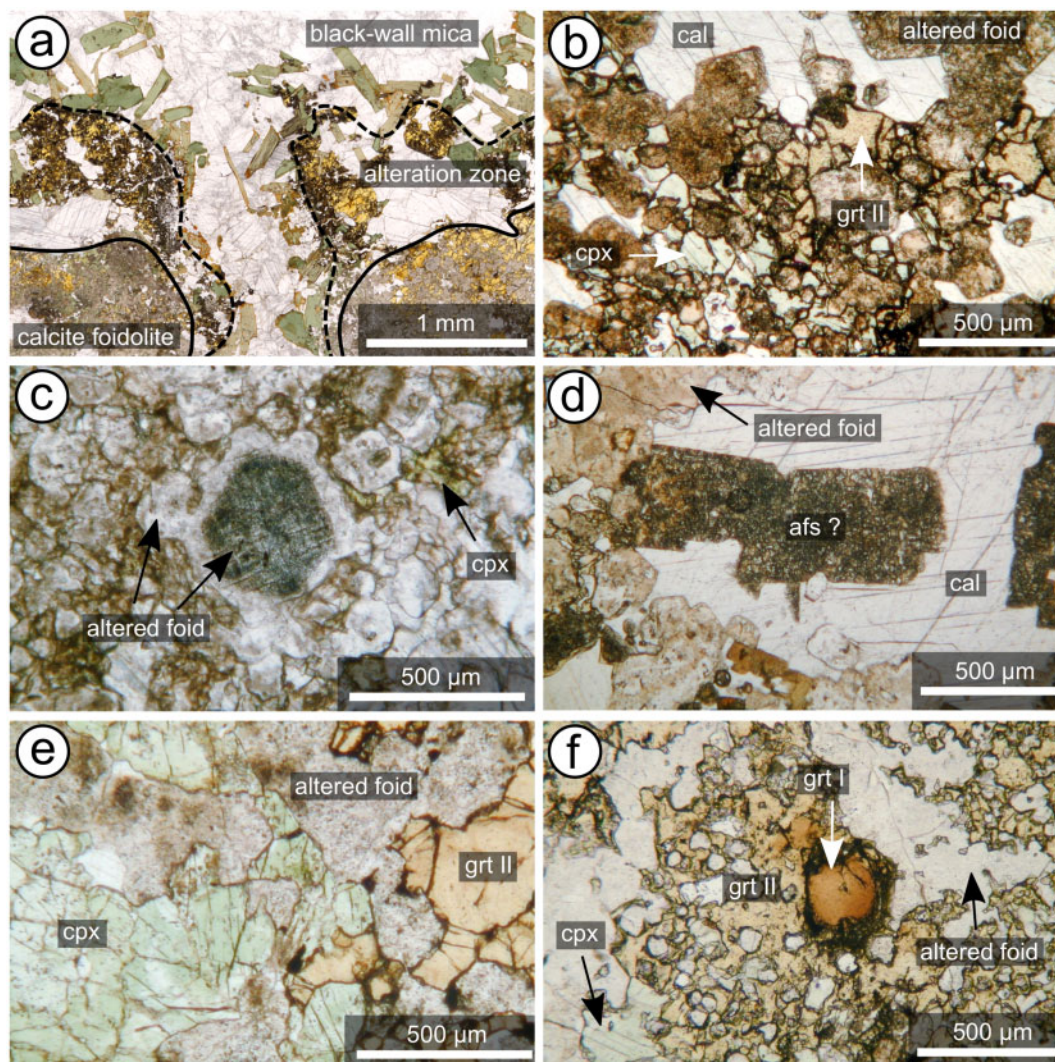


Fig. 7. Photomicrographs (plane polarized light) of the mineral assemblage and textural appearance of calcite foidolites from the KVC. (a) Coarse-grained biotite-rich ‘black-wall’ at the contact between carbonatite and calcite foidolite. (b) Recrystallized garnet (II) associated with recrystallized clinopyroxene (II), calcite and foid minerals (mostly altered to zeolites). (c) Euhedral nosean (or haüyne) decomposed to zeolite and calcite. (d) Alkali feldspar (afs) replaced by a mixture of zeolite, calcite, biotite, minor clinopyroxene (II) and garnet (II). (e) Clinopyroxene (II), garnet (II) and interstitial calcite in a calcite foidolite. (f) Dark brown and rounded (relict) garnet (I) overgrown by anhedral garnet (II).

Clinopyroxene occurs mostly as pale green, sometimes patchy-coloured subhedral and interstitial grains (Fig. 7e). If present, dark brown garnet I (relict cores) is overgrown by a reddish-brownish garnet II (recrystallized anhedral grains or masses; Fig. 7f). Garnet II is commonly zoned and occasionally forms poikilitic aggregates that enclose former foid minerals (Fig. 7f). Ubiquitous accessory minerals are apatite and titanite, only one sample contains primary spinel group minerals.

MINERAL COMPOSITIONS

Carbonates

Representative analyses are given in Table 2; formula calculations are based on one cation with CO₂ being calculated. Calcite in sövites contains variable, but

generally minor amounts of Sr (<2 wt % SrO), Fe (<0.4 wt % FeO), Mg (<1.5 wt % MgO) and Mn (<1.5 wt %). Mean REE contents are generally <0.2 wt % Ce₂O₃ (similar to earlier data; Hornig-Kjarsgaard, 1998), but may occasionally reach higher values; however, the presence of REE-rich micro inclusions cannot be excluded. Calcite from Badberg samples is relatively Sr-rich but Mg-poor compared to calcite from other sövites (Fig. 8). Calcite in extrusive carbonatites is highly variable in composition. In the crystal tuff and lapillistone samples from Henkenberg, primary calcite is Sr-rich but Mg-poor compared to secondary calcite cement (Fig. 8).

Apatite

Representative analyses are given in Table 3 and formula calculations were done using the method of Ketcham (2015). As typical of carbonatites

Table 2: Representative EPMA analyses of carbonates from the Kaiserstuhl Volcanic Complex

Sample #	1229	1237	1362	1267	1294	1318	1316b	1410	237	239	225	1356	1366
Locality	Badberg			Orberg II				Orberg III	Orberg IV		Orberg V		
Mineral	Calcite			Dolomite			Ankerite			Calcite			
Rock type	Sövite												
wt.%	55.18	55.45	55.09	55.79	54.74	28.89	28.88	53.15	54.88	54.67	54.39	56.22	55.63
CaO	0.08	0.04	0.03	0.40	0.12	18.57	8.88	1.26	0.50	0.58	0.55	0.05	0.36
MgO	bdl	0.04	bdl	0.04	0.16	0.53	15.15	bdl	0.11	bdl	0.22	0.14	0.26
FeO	0.10	0.24	0.03	bdl	0.38	2.38	3.21	0.50	0.04	0.19	bdl	bdl	bdl
MnO	1.62	1.55	1.36	1.18	1.40	1.37	0.27	1.46	1.16	1.07	1.18	0.84	0.00
SrO	0.07	0.89	0.07	0.21	0.11	0.04	bdl	bdl	0.64	0.06	bdl	0.06	bdl
Ce ₂ O ₃	44.17	44.75	43.89	44.83	44.07	45.35	43.75	44.01	44.46	44.13	43.92	44.64	44.21
CO ₂ (calc.)	101.22	102.95	100.46	102.44	100.98	97.13	100.14	100.38	101.78	100.70	100.25	101.95	100.46
Total													
Formula based on 1 cation													
Ca	0.98	0.97	0.99	0.98	0.97	0.50	0.52	0.95	0.97	0.97	0.97	0.99	0.99
Mg	0.002	0.001	0.001	0.01	0.003	0.37	0.22	0.03	0.01	0.01	0.01	0.001	0.01
Fe	bdl	0.00	bdl	0.00	0.002	0.05	0.21	bdl	0.002	bdl	0.003	0.002	0.004
Mn	0.001	0.003	0.00	bdl	0.01	0.04	0.05	0.01	0.001	0.003	bdl	bdl	bdl
Sr	0.02	0.01	0.01	0.01	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.00
Ce	0.001	0.008	0.001	0.002	0.001	0.0003	bdl	bdl	0.01	0.001	bdl	0.001	bdl
Total cations	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00
Sample #	284	1354	1399	1415b	1415c	GM1	GM2	GM4	GM4	GM4	1337	1337	1337
Locality	Haselschacher Buck		Degenmatt	Katharinenberg		Henkenberg					Kirchberg		
Mineral	Calcite			Crystal tuff			Lapilli stone			Sec. Calcite			
Rock type	Sövite									Lava			
Wt %	54.86	54.93	51.87	55.31	53.13	56.44	55.03	56.35	56.14	56.65	54.82	55.65	55.65
CaO	0.31	0.18	0.48	0.53	1.19	0.43	0.05	0.08	0.82	0.04	0.03	0.04	0.04
MgO	0.29	bdl	0.13	0.09	0.07	bdl	bdl	0.06	bdl	bdl	0.40	bdl	bdl
FeO	bdl	bdl	1.17	0.14	0.46	bdl	0.15	0.02	bdl	bdl	bdl	bdl	bdl
MnO	0.87	0.72	0.26	0.38	1.12	0.00	1.48	1.00	bdl	bdl	1.36	1.15	1.15
SrO	bdl	bdl	0.06	0.17	0.09	bdl	0.09	0.08	0.05	bdl	bdl	bdl	bdl
Ce ₂ O ₃	43.94	43.69	42.18	44.36	43.83	44.76	44.00	44.82	44.97	44.21	43.88	44.21	44.21
CO ₂ (calc.)	100.28	99.65	96.14	100.99	99.87	101.63	100.80	102.41	101.98	101.04	100.49	101.04	101.04
Total													
Formula based on 1 cation													
Ca	0.98	0.99	0.97	0.98	0.95	0.99	0.98	0.99	0.98	0.99	0.98	0.99	0.99
Mg	0.01	0.004	0.01	0.01	0.03	0.01	0.001	0.002	0.02	0.001	0.001	0.001	0.001
Fe	0.004	bdl	0.002	0.001	0.001	bdl	bdl	0.001	bdl	bdl	0.01	bdl	bdl
Mn	bdl	0.002	0.02	0.002	0.01	bdl	0.002	0.00	bdl	bdl	bdl	bdl	bdl
Sr	0.01	0.01	0.00	0.00	0.01	0.00	0.01	0.01	bdl	bdl	0.01	0.01	0.01
Ce	bdl	bdl	0.001	0.002	0.001	bdl	0.001	0.001	0.001	0.000	bdl	bdl	bdl
Total cations	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00

bdl, below detection limit.

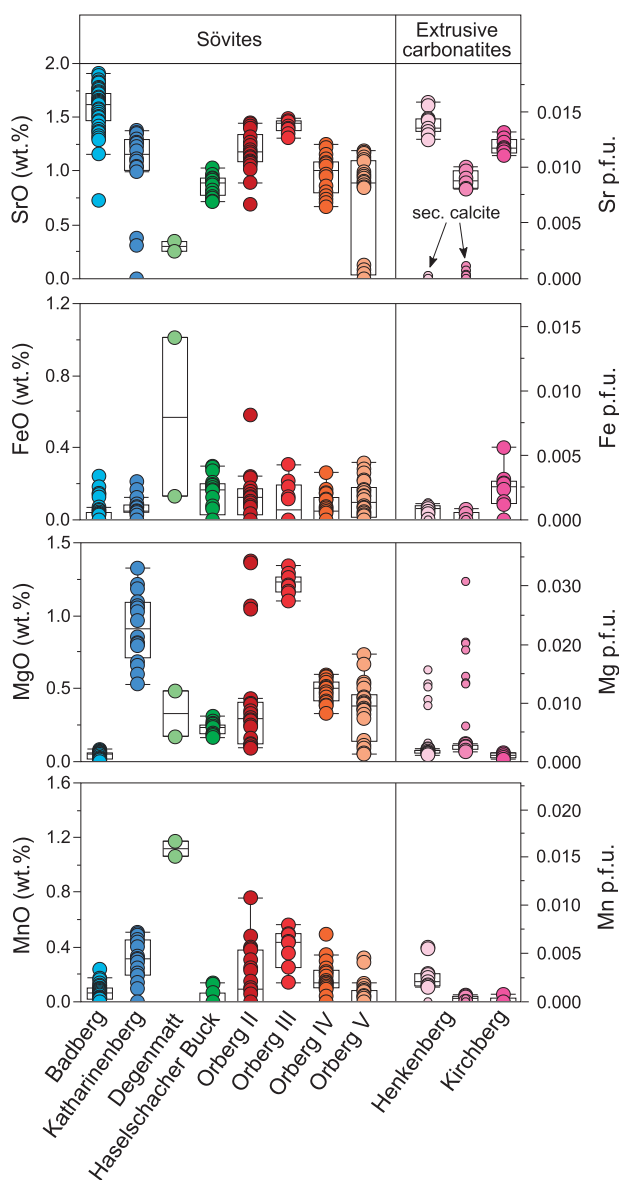


Fig. 8. Composition of calcite in sövites and extrusive carbonatites from the KVC shown as box plot diagrams.

(Chakhmouradian *et al.*, 2017), apatite from the KVC is fluorapatite to hydroxyapatite with variable F (1–2.8 wt %) and very low Cl contents (below 0.02 wt %; Table 4). Most data closely follow a substitution mechanism according to the apatite-britholite series ($\text{Ca}^{2+} + \text{P}^{5+} \leftrightarrow \text{REE}^{3+} + \text{Si}^{4+}$; Fig. 9a). Apatites from the Badberg sövites and calcite foidolites reach the highest levels of REE_2O_3 (up to 6.5 wt %), SrO (up to 4 wt %) and SiO_2 (up to 3.6 wt %), but are relatively low in Na_2O (below 0.5 wt %) compared to other locations (Fig. 9). In contrast, apatites from Orberg are low in REE_2O_3 (<3 wt %), SiO_2 (<1.8 wt %) and SrO (<1.5 wt %), but in some cases show elevated Na_2O (up to 1.2 wt %), following the coupled substitution mechanism $2\text{Ca}^{2+} + (3\text{Ca}^{2+}) \leftrightarrow \text{REE}^{3+} + \text{Na}^+ + (3\text{Sr}^{2+})$ (apatite-belovite series). Most apatites are compositionally zoned with variable enrichments of REE and Si towards the rim of the crystals. In

particular, apatites from the Badberg sövite show a distinct and sharp enrichment of a britholite component (Figs 3d and 9a) and commonly a slight decrease of Sr in their rims (Fig. 9c). Apatite data from samples of the KB3 drill hole (Badberg) show greater scatter in the upper parts of the drill hole than in the lower parts, with higher REE and Si but lower F and slightly lower Sr in the deeper parts of the drill hole (Fig. 10).

Mica

Representative analyses are given in Table 4; formula calculations are based on an ideal trioctahedral mica formula $(\text{XY}_3[\text{Z}_4\text{O}_{10}][\text{OH}, \text{F}, \text{Cl}]_2)$ and the data were normalized to 7 (tetrahedral [Z] plus octahedral [Y]) cations. Micras from the KVC are Mg-rich and show variable enrichment of Ba (Fig. 11a), with those from Badberg having lower Ba (<0.07 apfu) and Mg# (72–88) compared to micras from the other localities (up to 0.5 Ba pfu and Mg# 91–97), while tetrahedral Fe^{3+} is similarly low at all localities (<0.05 and <0.07 Fe^{3+} pfu, respectively). Accordingly, mica from Badberg is classified as biotite (after Foster, 1960), whereas mica from all other localities is phlogopite. A continuous 1:1 increase of Al with Ba is caused by the phlogopite-kinoshitalite substitution ($\text{K}^+ + \text{Si}^{4+} \leftrightarrow \text{Ba}^{2+} + \text{IVAl}^{3+}$). Deviations from the 1:1 slope towards higher Al indicate an additional effect of the phlogopite-eastonite substitution ($\text{Mg}^{2+} + \text{Si}^{4+} \leftrightarrow \text{IVAl}^{3+} + \text{VIAl}^{3+}$; Fig. 11b) with a shift from the ideal 1:1 line reflecting constant eastonite substitution with increasing kinoshitalite substitution. The distinct oscillatory zonation in phlogopites from Orberg III and Katharinenberg (Fig. 4b) exclusively reflects a variable kinoshitalite substitution. Based on specific chemical characteristics, three mica groups are distinguished:

1. Phlogopites from Degenmatt, Katharinenberg and Orberg III and V are characterized by a large spread in Ba and Al contents (Fig. 11b). Orberg III micras follow a nearly ideal kinoshitalite substitution, whereas for micras from Degenmatt, Katharinenberg and Orberg V the additional importance of the eastonite substitution is obvious (Fig. 11a). Phlogopites of this group are further characterized by intermediate Na and F (Na up to 0.09 apfu, F up to 0.4 apfu; Fig. 11c and d) and the lowest Ti and Mn contents (<0.02 and <0.03 apfu, respectively; Fig. 11e and f).
2. Phlogopites from Orberg II and IV and from Haselschacher Buck show a constant eastonite substitution with increasing kinoshitalite substitution (Ba <0.3 apfu; Fig. 11b) and are further characterized by the highest Na (up to 0.11 apfu), intermediate Ti, (up to 0.03 apfu) and low Mn (<0.03 apfu) and F (<0.15 apfu; Fig. 11c–f) contents.
3. Biotite from the Badberg shows the lowest Mg# (72–88) and although their Ba content is low (<0.07 apfu), they show strong Al enrichment during evolution suggesting the importance of the eastonite substitution (Fig. 11a and b). They are further characterized by the highest Mn (up to 0.1 apfu), low

Table 3: Representative EPMA analyses of apatite from the Kaiserstuhl Volcanic Complex

Sample #	1229	1139	1147	1178	1253	1413	1410	1353	239	237	222	222	225	1356	1399	284	1354	1415c	GM3																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Locality	Badberg	Badberg		Orberg II	Orberg III	Orberg III	Orberg III	Orberg IV	Orberg IV	Orberg V	Orberg V	Orberg V	225	1356	Degenmatt	Haselschacher Buck	1354	Katharinenberg	Henkenberg																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Rock type	Calcite Foidolite										Sövite										Crystal tuff																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																							
wt %																					Na ₂ O	0.08	0.10	0.06	0.09	0.22	0.25	0.25	0.21	0.31	0.17	1.06	0.69	0.29	0.04	0.51	0.13	0.12	0.18	0.14	CaO	52.25	52.04	51.00	51.98	54.13	54.31	54.39	54.10	54.00	53.46	52.47	52.38	53.55	52.72	51.31	53.88	53.06	54.82	52.79	SrO	0.90	1.48	1.94	1.25	0.62	0.85	0.81	0.76	0.61	0.67	1.41	1.40	0.68	0.53	1.47	0.49	0.46	0.56	0.51	La ₂ O ₃	1.31	0.60	1.41	0.92	0.57	0.24	0.28	0.25	0.41	0.25	0.40	0.42	0.31	0.53	0.37	0.64	0.66	0.21	0.16	Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01
Na ₂ O	0.08	0.10	0.06	0.09	0.22	0.25	0.25	0.21	0.31	0.17	1.06	0.69	0.29	0.04	0.51	0.13	0.12	0.18	0.14	CaO	52.25	52.04	51.00	51.98	54.13	54.31	54.39	54.10	54.00	53.46	52.47	52.38	53.55	52.72	51.31	53.88	53.06	54.82	52.79	SrO	0.90	1.48	1.94	1.25	0.62	0.85	0.81	0.76	0.61	0.67	1.41	1.40	0.68	0.53	1.47	0.49	0.46	0.56	0.51	La ₂ O ₃	1.31	0.60	1.41	0.92	0.57	0.24	0.28	0.25	0.41	0.25	0.40	0.42	0.31	0.53	0.37	0.64	0.66	0.21	0.16	Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																					
CaO	52.25	52.04	51.00	51.98	54.13	54.31	54.39	54.10	54.00	53.46	52.47	52.38	53.55	52.72	51.31	53.88	53.06	54.82	52.79	SrO	0.90	1.48	1.94	1.25	0.62	0.85	0.81	0.76	0.61	0.67	1.41	1.40	0.68	0.53	1.47	0.49	0.46	0.56	0.51	La ₂ O ₃	1.31	0.60	1.41	0.92	0.57	0.24	0.28	0.25	0.41	0.25	0.40	0.42	0.31	0.53	0.37	0.64	0.66	0.21	0.16	Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																									
SrO	0.90	1.48	1.94	1.25	0.62	0.85	0.81	0.76	0.61	0.67	1.41	1.40	0.68	0.53	1.47	0.49	0.46	0.56	0.51	La ₂ O ₃	1.31	0.60	1.41	0.92	0.57	0.24	0.28	0.25	0.41	0.25	0.40	0.42	0.31	0.53	0.37	0.64	0.66	0.21	0.16	Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																													
La ₂ O ₃	1.31	0.60	1.41	0.92	0.57	0.24	0.28	0.25	0.41	0.25	0.40	0.42	0.31	0.53	0.37	0.64	0.66	0.21	0.16	Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																	
Ce ₂ O ₃	2.07	1.18	1.98	1.49	1.13	0.30	0.43	0.47	0.65	0.50	0.67	0.71	0.57	1.37	0.60	1.10	1.14	0.51	0.37	Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																					
Pr ₂ O ₃	0.27	0.19	0.25	0.16	0.12	bdl	0.07	0.07	0.13	bdl	0.06	0.16	0.04	0.27	0.03	0.15	0.16	bdl	bdl	Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																									
Nd ₂ O ₃	0.46	0.24	0.39	0.30	0.47	0.09	0.15	bdl	0.28	0.12	0.24	0.26	0.18	0.42	0.25	0.38	0.39	0.17	0.12	FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																													
FeO	bdl	0.03	0.04	0.03	bdl	1.07	0.02	bdl	bdl	bdl	0.02	0.07	bdl	bdl	0.04	bdl	0.15	bdl	0.23	P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																	
P ₂ O ₅	36.62	39.83	36.73	38.86	40.20	39.22	40.72	40.30	40.01	41.19	39.64	39.78	40.26	39.27	40.86	38.48	38.85	40.38	38.77	SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																					
SiO ₂	2.55	0.87	2.50	1.29	0.84	0.21	0.33	0.48	0.52	0.35	0.02	0.17	0.72	1.50	0.04	1.62	1.92	0.53	0.98	SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																									
SO ₃	0.38	0.08	0.34	0.12	0.14	0.17	0.24	0.27	0.22	0.14	0.32	0.24	0.32	0.20	0.12	0.44	0.24	0.23	0.97	As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																													
As ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	bdl	0.03	bdl	bdl	bdl	bdl	0.04	bdl	bdl	bdl	MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																	
MnO	bdl	0.04	bdl	bdl	0.02	0.08	0.05	0.02	bdl	0.02	0.12	0.08	0.04	bdl	0.07	bdl	bdl	0.04	0.03	Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																					
Cl	0.05	0.05	0.02	0.01	0.03	0.03	0.03	0.02	0.05	0.03	0.02	0.04	0.05	0.01	0.01	0.07	0.03	0.04	0.62	F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																									
F	1.41	2.51	1.09	2.08	1.62	1.70	1.66	1.44	1.42	1.71	1.93	2.08	1.33	1.11	2.73	0.99	1.14	1.45	2.30	–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																													
–O(II)	0.60	1.06	0.46	0.88	0.68	0.72	0.70	0.61	0.60	0.72	0.81	0.87	0.56	0.47	1.15	0.42	0.48	0.61	0.97	–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																	
–O(Cl)	0.01	0.01	0.00	0.00	0.01	0.01	0.01	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.02	0.01	0.01	0.14	Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																					
Corrected Total:	97.76	98.17	97.28	97.70	99.42	97.78	98.72	97.79	98.02	97.87	97.56	97.58	97.76	97.75	97.26	97.94	97.81	98.49	96.90	*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																									
*OH wt. %	1.93	0.99	2.23	1.38	1.83	1.72	1.80	1.99	1.99	1.77	1.52	1.37	2.08	2.27	0.82	2.35	2.24	1.98	0.91	Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																													
Net Cor. Total:	99.69	99.16	99.51	99.08	101.25	99.50	100.52	99.78	100.01	99.64	99.08	98.95	99.84	100.02	98.08	100.29	100.05	100.47	97.81	Formula calculation based on the method of Ketchum (2015)																				Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																	
Formula calculation based on the method of Ketchum (2015)																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																												
Na	0.01	0.02	0.01	0.01	0.04	0.04	0.04	0.04	0.05	0.03	0.18	0.12	0.05	0.01	0.09	0.02	0.02	0.03	0.02	Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Ca	4.92	4.83	4.82	4.86	4.94	5.05	4.96	4.97	4.97	4.89	4.89	4.88	4.91	4.87	4.77	4.97	4.89	5.00	4.92	Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																													
Sr	0.05	0.07	0.10	0.06	0.03	0.04	0.04	0.04	0.03	0.03	0.07	0.07	0.03	0.03	0.07	0.02	0.02	0.03	0.03	La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																	
La	0.04	0.02	0.05	0.03	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.02	0.02	0.01	0.01	Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																					
Ce	0.07	0.04	0.06	0.05	0.04	0.01	0.01	0.01	0.02	0.02	0.02	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Pr	0.01	0.01	0.01	0.01	0.00	bdl	0.00	0.00	0.00	bdl	0.00	0.01	0.00	0.01	0.00	0.00	0.00	bdl	bdl	Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																													
Nd	0.01	0.01	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.00	Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																	
Fe	bdl	0.00	0.00	0.00	bdl	0.08	0.00	0.00	bdl	bdl	0.00	0.00	bdl	bdl	0.00	bdl	0.01	bdl	0.02	P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																					
P	2.72	2.92	2.75	2.87	2.90	2.88	2.93	2.92	2.91	2.98	2.92	2.93	2.92	2.87	3.00	2.81	2.83	2.91	2.85	Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Si	0.22	0.08	0.22	0.11	0.07	0.02	0.03	0.04	0.04	0.03	0.00	0.01	0.06	0.13	0.00	0.14	0.16	0.05	0.09	S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																													
S	0.03	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.03	0.02	0.01	0.06	As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																	
As	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	bdl	0.00	bdl	bdl	bdl	bdl	0.00	bdl	bdl	bdl	Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																					
Mn	bdl	0.00	bdl	bdl	0.00	0.01	0.00	0.00	bdl	0.00	0.01	0.01	0.00	bdl	0.01	bdl	bdl	0.00	0.00	Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									
Cl	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.01	0.09	F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																													
F	0.39	0.69	0.30	0.57	0.44	0.47	0.45	0.39	0.39	0.46	0.53	0.57	0.36	0.30	0.75	0.27	0.31	0.39	0.63	OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																	
OH ^(est.)	0.60	0.30	0.69	0.42	0.56	0.53	0.55	0.61	0.61	0.53	0.47	0.42	0.63	0.69	0.25	0.72	0.69	0.60	0.28	Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																					
Total cations	8.08	8.00	8.06	8.03	8.05	8.15	8.05	8.05	8.07	7.99	8.14	8.09	8.03	8.00	8.00	8.07	8.03	8.06	8.01																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																									

*OH wt. %, calculated from OH^(est.), OH^(est.), estimated after calculations of [Ketchum \(2015\)](#).
 bdl, below detection limit. Noticeable variation of representative analyses in a single sample has been indicated by the addition of further analyses, reflecting the variation.

Table 4: Representative EPMA analyses of micas from the Kaiserstuhl Volcanic Complex

Sample # Locality	Sövite (Black Wall)										Sövite									
	1200 Badberg	1200	1156	1148	1245	1237	1318 Orberg I	1318 Orberg II	1413 Orberg III	1413	1410	237 Orberg IV	222 Orberg V	1366	1354 Haselschacher Buck	284	1415b Katharinenberg	1415c		
Rock type	Sövite (Black Wall)										Sövite									
wt %																				
Na ₂ O	0.23	0.23	0.22	0.29	0.36	0.41	0.17	0.13	0.40	0.50	0.58	0.52	0.64	0.29	0.51	0.42	0.66	0.66	0.48	
K ₂ O	9.69	10.16	9.89	9.26	9.39	9.33	9.81	10.63	4.62	5.25	6.77	7.16	7.33	7.93	6.58	8.90	8.23	6.66	7.46	
CaO	0.02	bdl	0.06	0.03	0.05	0.04	0.18	0.26	0.19	0.01	0.12	0.07	0.05	0.04	0.02	0.04	0.07	0.23	0.05	
BaO	1.74	0.24	0.82	2.42	2.09	2.29	2.56	0.44	14.76	12.57	8.12	8.11	6.81	7.44	10.15	3.21	4.21	8.66	7.20	
MgO	24.02	21.79	20.33	21.40	21.03	22.67	25.62	27.43	23.97	24.42	24.85	24.97	24.06	26.21	25.23	27.02	24.67	24.41	25.62	
MnO	1.03	1.36	1.17	1.08	0.75	0.69	0.36	0.40	0.73	0.17	0.21	0.23	0.35	0.20	0.14	0.26	0.29	0.30	0.11	
FeO ^s	5.20	9.25	10.46	7.40	8.82	7.71	2.24	1.89	0.73	1.78	1.86	2.22	4.21	1.60	1.30	1.36	3.72	2.72	1.60	
Fe ₂ O ₃	0.69	0.01	0.25	0.96	0.18	0.13	0.00	0.15	0.00	0.11	0.40	0.11	0.04	0.05	0.32	1.11	0.00	0.20	0.16	
Al ₂ O ₃	13.74	11.87	13.16	15.39	15.08	14.99	15.06	12.25	20.11	19.29	18.02	18.12	17.07	15.77	18.30	13.87	17.81	18.65	16.57	
TiO ₂	0.17	0.76	0.37	0.22	0.55	0.57	1.00	0.18	0.15	0.17	0.29	0.13	0.31	0.22	0.15	0.22	0.39	0.23	0.11	
Cr ₂ O ₃	bdl	0.05	0.02	bdl	bdl	0.01	0.01	bdl	bdl	0.03	0.05	0.03	0.02	0.01	bdl	bdl	0.01	bdl	0.03	
SiO ₂	39.51	40.27	39.10	37.38	37.90	38.85	38.70	42.68	29.77	31.95	34.40	35.33	36.01	37.17	34.34	39.20	37.13	34.63	36.49	
Cl	bdl	0.01	0.02	0.01	bdl	0.01	bdl	bdl	0.01	0.02	0.02	bdl	0.02	0.02	0.01	bdl	bdl	bdl	bdl	
F	2.09	1.26	1.32	1.27	0.08	0.19	1.58	3.10	1.14	0.80	0.85	0.93	0.69	1.55	0.88	1.41	0.17	0.15	0.79	
-O=(F, Cl) ₂	0.88	0.53	0.56	0.54	0.04	0.08	0.67	1.31	0.48	0.34	0.36	0.39	0.30	0.66	0.37	0.59	0.07	0.06	0.33	
Total	98.13	97.26	97.19	97.10	96.28	97.90	97.29	99.39	97.13	96.94	96.54	97.93	97.61	98.50	97.92	97.02	97.36	97.50	97.57	
H ₂ O*	3.17	3.51	3.44	3.47	4.05	4.09	3.43	2.82	3.34	3.57	3.64	3.67	3.77	3.39	3.65	3.52	4.12	4.02	3.65	
Formula based on 7 (tetrahedral [Z] plus octahedral [Y]) cations																				
Na	0.03	0.03	0.03	0.04	0.05	0.06	0.02	0.02	0.06	0.07	0.08	0.07	0.09	0.04	0.07	0.06	0.09	0.09	0.07	
K	0.89	0.94	0.93	0.87	0.88	0.85	0.90	0.95	0.46	0.51	0.64	0.67	0.68	0.73	0.62	0.81	0.75	0.62	0.69	
Ca	0.001	bdl	0.005	0.002	0.004	0.003	0.01	0.02	0.02	0.001	0.01	0.01	0.004	0.003	0.001	0.003	0.01	0.02	0.003	
Ba	0.05	0.01	0.02	0.07	0.06	0.06	0.07	0.01	0.45	0.37	0.24	0.23	0.20	0.21	0.29	0.09	0.12	0.25	0.32	
Mg	2.57	2.36	2.23	2.34	2.30	2.42	2.74	2.86	2.75	2.74	2.73	2.71	2.61	2.83	2.77	2.87	2.61	2.66	2.77	
Mn	0.06	0.08	0.07	0.07	0.05	0.04	0.02	0.02	0.01	0.01	0.01	0.01	0.02	0.01	0.01	0.02	0.02	0.02	0.01	
Fe ²⁺	0.31	0.56	0.64	0.45	0.54	0.46	0.13	0.11	0.05	0.11	0.11	0.14	0.26	0.10	0.08	0.08	0.22	0.17	0.14	
Fe ³⁺	0.04	0.00	0.01	0.05	0.01	0.01	0.00	0.00	0.07	0.00	0.02	0.01	0.00	0.00	0.02	0.06	0.00	0.01	0.01	
Al	1.16	1.02	1.14	1.33	1.30	1.26	1.27	1.01	1.83	1.71	1.57	1.55	1.47	1.35	1.59	1.17	1.49	1.61	1.42	
Ti	0.01	0.04	0.02	0.01	0.03	0.03	0.05	0.01	0.01	0.01	0.02	0.01	0.02	0.01	0.01	0.01	0.02	0.01	0.01	
Cr	bdl	0.003	0.001	bdl	bdl	0.001	0.001	bdl	bdl	0.001	0.003	0.002	0.001	0.001	bdl	bdl	0.001	bdl	0.001	
Si	2.84	2.93	2.88	2.74	2.78	2.78	2.78	2.99	2.29	2.41	2.54	2.57	2.62	2.69	2.53	2.79	2.64	2.53	2.48	
Cl	bdl	0.001	0.002	0.002	0.000	0.001	bdl	bdl	0.001	0.002	0.003	bdl	0.002	0.002	0.001	bdl	bdl	bdl	bdl	
F	0.48	0.29	0.31	0.30	0.02	0.04	0.36	0.68	0.28	0.19	0.20	0.21	0.16	0.36	0.21	0.32	0.04	0.04	0.19	
Total cations	7.97	7.97	7.97	7.98	7.99	7.98	8.01	8.00	7.98	7.96	7.97	7.98	7.97	7.99	7.99	7.96	7.96	7.98	7.97	

H₂O*, calculated from OH (F+Cl+OH = 2). S, Fe²⁺/Fe³⁺ ratios calculated from stoichiometry.

bdl, below detection limit. Noticeable variation of representative analyses in a single sample has been indicated by the addition of further analyses, reflecting the variation.

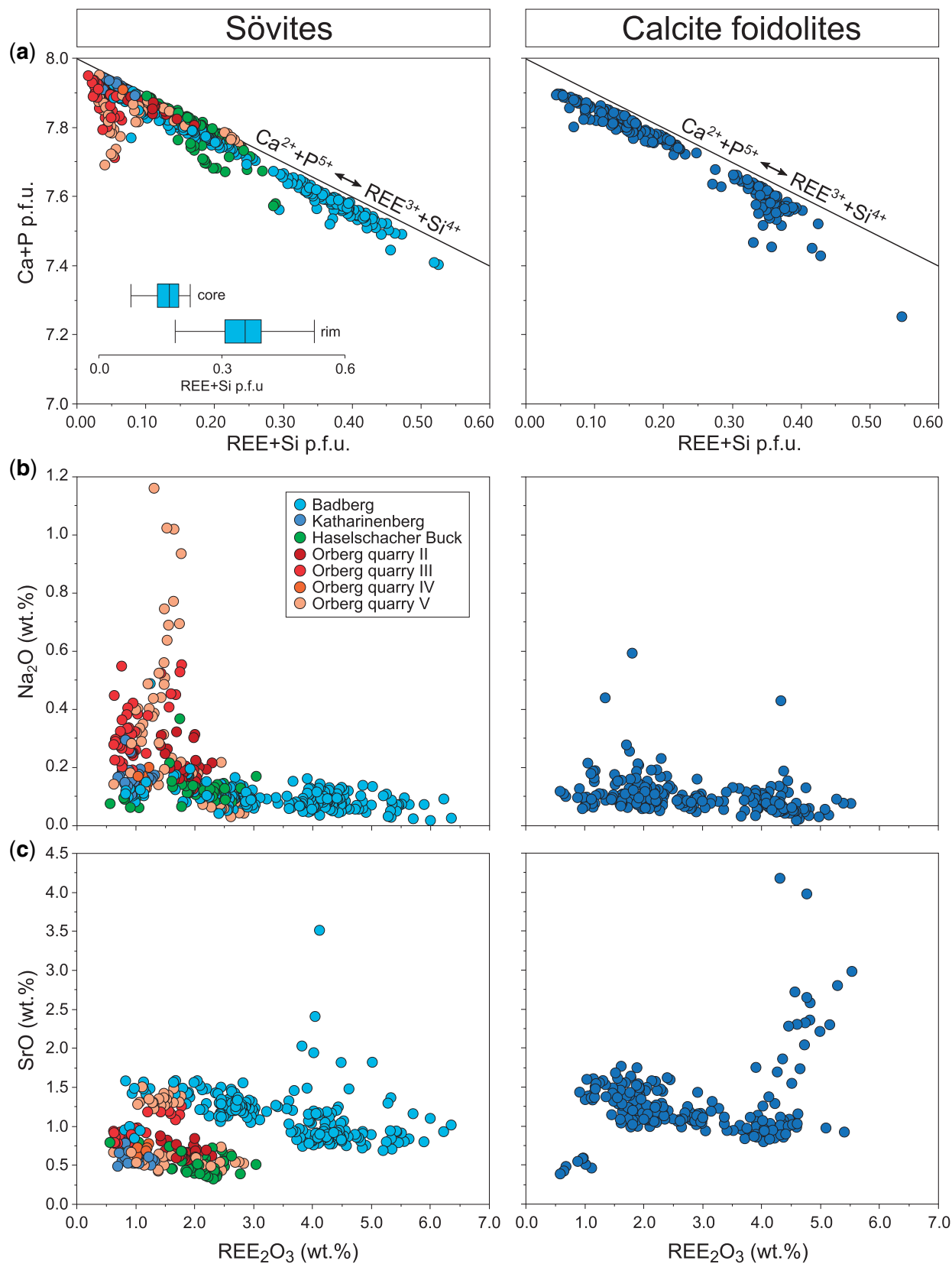


Fig. 9. Chemical variations of apatite in sövites and calcite foidolites from the KVC. (a) Ca+P vs REE+Si representing the apatite-britholite substitution mechanism ($Ca^{2+} + P^{5+} \leftrightarrow REE^{3+} + Si^{4+}$). (b) and (c) Variations of Na₂O and SrO vs REE₂O₃.

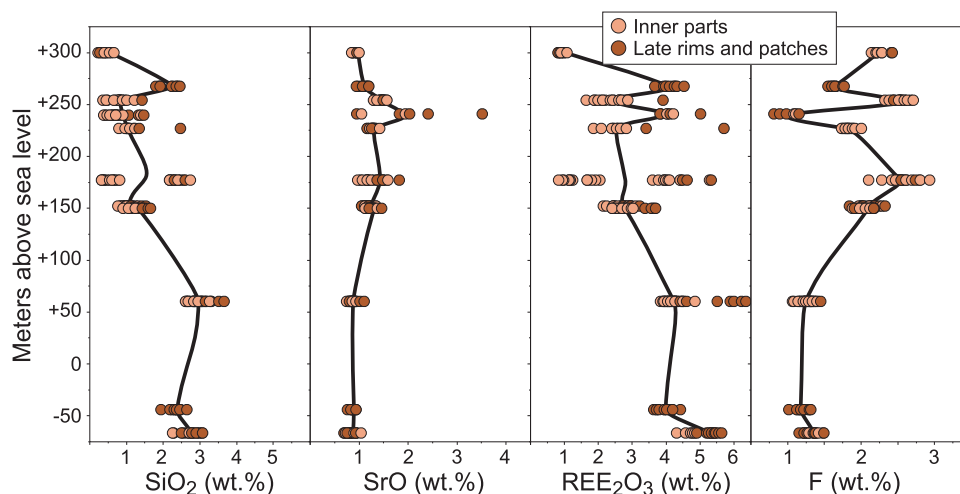


Fig. 10. Compositional variations of apatite with depth (given as m above sea level) in the KB3 drill hole (Badberg). Note: According to the tilting of the Badberg sövite (Fig. 1c), this variation depicts only a restricted depth variation within the sövite pipe, but reflects an additional centre-margin effect.

to intermediate Na (<0.07 apfu) and F (<0.5 apfu) and intermediate to high Ti (<0.06 apfu; Fig. 11c–f). Amongst the Badberg micas, those from black-walls around calcite foidolites (Fig. 7a) show lower Ti and Na, but higher F and Mn contents (Fig. 11c–f), but are otherwise compositionally similar to those formed far away from xenoliths.

Olivine and monticellite

Representative analyses are given in Table 5; formula calculations are based on four oxygens. Olivine in two of the sövites is Mg-rich ($F_{0.96-0.90}$) but shows a large range in the larnite component, varying from 0.1 to 1.5 mol % (0.06–1.17 wt % CaO; Fig. 12a). The negative correlation between forsterite and larnite components (Fig. 12a) implies a preferred incorporation of Ca with decreasing Mg (Jurewicz & Watson, 1988; Libourel, 1999). The Mn contents (up to 3.0 wt % MnO) are much higher than in the associated silicate rocks of the KVC (grey fields in Fig. 12b), while Ni contents rarely exceed the detection limit (NiO <0.03 wt %). Monticellite (about 75 mol %) is present in one sample (Orberg V) and contains about 20 mol % kirschsteinite (CaFeSiO_4) and 5 mol % glaucochroite (CaMnSiO_4).

Clinopyroxene

Formula calculations are based on four cations and six oxygens; representative analyses are listed in Table 6. Clinopyroxene in the investigated samples is invariably diopside-rich (Fig. 12c and d). However, clinopyroxene in sövites and calcite foidolites is higher in MnO (up to 1.4 wt %) but lower in TiO_2 (<0.9 wt %) compared to clinopyroxene in extrusive carbonatites (MnO <0.3 wt % and up to 4.1 wt % TiO_2 ; Fig. 12c and d). The concentration of Cr_2O_3 (up to 0.27 wt %) is only elevated in some extrusive carbonatites (Table 6).

Spinel group minerals

Representative analyses are given in Table 7 and formula calculations are based on three cations and four oxygens. Spinel group minerals in the investigated samples are classified as magnetite–magnesioferrite. In sövites they are poor in Ti (<0.13 apfu) and Al (<0.32 apfu), but relatively rich in Mn (up to 0.22 apfu), compared to those from extrusive carbonatites (Fig. 12e). In extrusive carbonatites, their Mg contents are relatively constant (0.26 to 0.36 apfu) compared to the large range observed in the sövite samples (0.08–0.65 apfu). Amongst the sövites, samples from Orberg reach the highest Mg content, implying a relatively large magnesioferrite component (up to 63 mol %), when considering endmembers magnetite ($\text{Fe}^{2+}\text{Fe}_2^{3+}\text{O}_4$), magnesioferrite ($\text{MgFe}_2^{3+}\text{O}_4$) and jacobsonite ($\text{MnFe}_2^{3+}\text{O}_4$). Spinel group mineral rims in a carbonatitic lava sample (Fig. 5b) that appear distinctly darker in BSE have high Mg (0.63–0.69 apfu) and Al (0.42–0.56 apfu) concentrations, i.e. are spinel-rich.

Garnet

Representative analyses are listed in Table 8; formula calculations are based on eight cations and twelve oxygens, using the calculation scheme of Locock (2008). Most analyses show a very good 1:1 correlation between $\text{Ti} + \text{Fe}^{3+}$ and $\text{Si} + \text{Al}$ (Fig. 13a), indicating solid solution between grossular (0–26%) andradite (47–84%), schorlomite (0–18%) and morimotoite (0–29%) endmembers (Fig. 13b). These garnets are best described as Ti-bearing andradites. They are commonly called ‘melanite’, but this name is not accepted by the IMA. The contents of Nb, Zr and REE are generally low (<0.02, <0.02 and <0.01 apfu, respectively) but garnet II from the calcite foidolites may reach much higher Nb (up to 0.2 apfu) and Zr (up to 0.2 apfu), whilst REE contents are similarly low (Fig. 13c–e). Such Nb- and Zr-rich

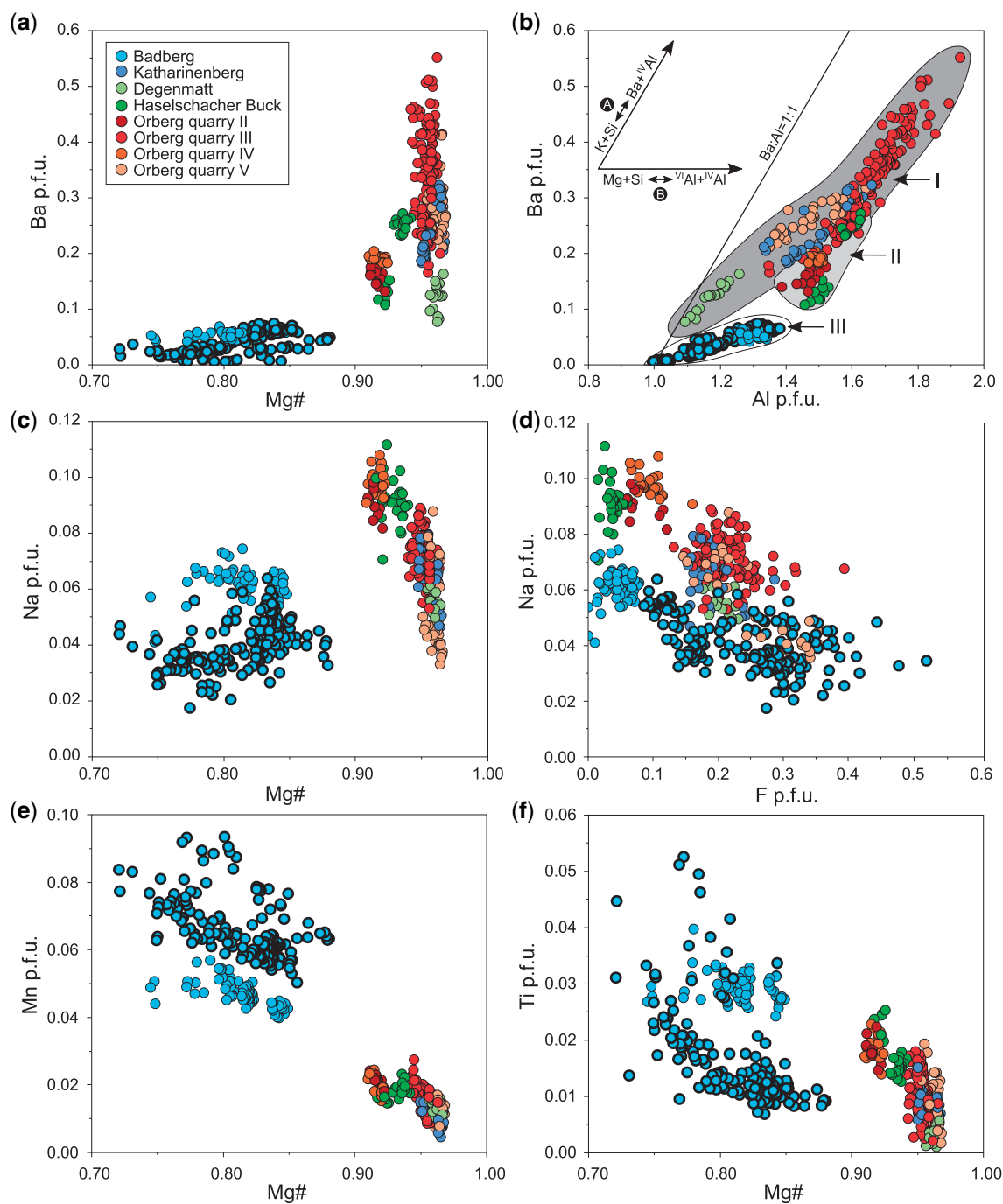


Fig. 11. Compositional variation of mica in sövites from the KVC. (a) Binary diagram of Ba vs Mg#. (b) Binary diagram of Ba versus Al. Inset depicts substitution tendencies: (A) Kinoshitalite substitution, (B) eastonite substitution (for details see text). (c) and (d) Na vs Mg# and F, respectively. (e) and (f) Mn and Ti vs Mg# respectively. Note the separation of biotite from Badberg compared to phlogopite from the other localities, as well as a separation between ‘black-wall’ mica (marked by heavier outline) and mica which is seemingly ‘independent’ of the presence of calcite foidolite in the Badberg samples.

analyses show clear deviations from the ideal $\text{Ti} + \text{Fe}^{3+} - \text{Si} + \text{Al}$ 1:1 correlation (Fig. 13a). Based on their Mg–Ti characteristics, garnet I and II from the calcite foidolites can be clearly distinguished from each other, with garnet from the extrusive carbonatites plotting in between these two groups (Fig. 13f).

Zirconolite

Representative analyses are given in Table 9; formula calculations are based on four cations. We analysed zirconolite in a sövite from Haselschacher Buck, where it is rich in Nb (up to 21.5 wt % Nb_2O_5) and Ta (up to 4.4 wt % Ta_2O_5) with relatively low amounts of REE+Y (up to

Table 5: Representative EPMA analyses of olivine and monticellite from the Kaiserstuhl Volcanic Complex

Sample #	1415	1415	1354	1354	1356
Locality	Katharinenberg		Haselschacher Buck		Orberg V
Mineral	Forsterite				Monticellite
wt. %					
SiO ₂	43.01	42.54	41.64	42.27	38.06
FeO	1.78	4.13	6.29	5.52	8.45
MnO	1.88	2.82	2.97	2.67	2.41
MgO	54.28	50.98	48.77	50.05	18.98
NiO	bdl	bdl	bdl	bdl	bdl
CaO	0.06	0.52	0.86	0.41	32.82
Total	101.03	100.99	100.53	100.92	100.74
Formula based on 4 oxygens					
Si	1.01	1.02	1.01	1.02	1.02
Fe	0.04	0.08	0.13	0.11	0.19
Mn	0.04	0.06	0.06	0.05	0.05
Mg	1.90	1.82	1.77	1.79	0.76
Ni	bdl	bdl	bdl	bdl	bdl
Ca	0.00	0.01	0.02	0.01	0.95
Total cations	2.99	2.98	2.99	2.98	2.98

bdl, below detection limit.

4.5 wt % REE₂O₃ + Y₂O₃, U (up to 1.2 wt % UO₂) and Th (up to 1.7 wt % ThO₂), plotting in the Nb+Ta-rich corner of the known compositional range (Fig. 14).

DISCUSSION

Origin of silicate minerals in the Kaiserstuhl carbonatites

Many carbonatites contain minor amounts of silicate minerals (in particular ferromagnesian silicates) with mica being the most common one (Reguir *et al.*, 2012). Their stability largely depends on silica activity, which is generally very low in carbonatites, in addition to temperature and other chemical potentials (Barker, 2001; Massuyeau *et al.*, 2015). Therefore, it is important to distinguish different genetic types of silicate minerals in carbonatites (e.g. Barker, 2001), including:

1. Silicates that crystallized from the carbonatitic magma, either due to a pristine sufficiently high silica activity ('true' primary magmatic silicates; type Ia) or because of a (local) change in silica activity (and the activity of other chemical components), for example by the interaction with wall rocks or xenoliths (type Ib).
2. Entrained crystals from the source rock or from wall rocks encountered during ascent and emplacement of the carbonatitic magma, including crystals from genetically related magmatic silicate rocks (xenocrysts; type II).
3. Subsolidus phases that formed during final cooling due to the interaction with hydrothermal or meteoric fluids (type III).

While a subsolidus origin (type III) can be easily recognized based on characteristic textures (e.g. exsolutions, pseudomorphs, veinlets), it is not easy to

distinguish primary (carbonatitic) crystals (type I) from xenocrysts (type II) because there is a broad range of compositional variations of these phases in carbonatites (Chakhmouradian & Zaitsev, 2002; Reguir *et al.*, 2009; Reguir *et al.*, 2012). Also, depending on their composition and reaction kinetics, entrained xenocrysts may or may not be partly resorbed and/or chemically modified by diffusion-controlled reactions with the carbonatite magma. Therefore, only careful textural analyses combined with a detailed comparison between the composition of the mineral in question in the carbonatite with that of mantle rocks (e.g. olivine) and surrounding silicate rocks may provide the opportunity to determine the origin of such silicate minerals. In the following, we discuss the potential origin of the various silicate phases present in the Kaiserstuhl carbonatites.

Mica is the major K-bearing phase in most carbonatites and is typically Fe²⁺-poor and variably enriched in Ba and Al (Giebel *et al.*, 2019 and references therein). At the KVC, mica in all sövites, except for the Badberg samples, shares these characteristics (Figs 11 and 15) and occurs as a minor phase more or less evenly dispersed in the rocks (Fig. 4a). We suggest that mica in these samples crystallized from the carbonatitic magma itself as a 'true' primary magmatic silicate at sufficiently high silica activities (type Ia). Mica in the Badberg samples, however, is compositionally distinct (Figs 11 and 15), much more abundant compared to the other sövites and partly occurs as seams around xenoliths (Figs 4 and 6). We assume that mica in the Badberg samples formed due to the interaction between entrained xenoliths and the carbonatitic magma (discussed in more detail below) and represents a type Ib phase.

Monticellite is stable at very low silica activities only (Barker, 2001) and is, therefore, commonly found in carbonatites (e.g. Nesbitt & Kelly, 1977; Stoppa & Lupini, 1993; Guzmics *et al.*, 2011), but is exceedingly rare in silicate rocks. None of the silicate lithologies at the KVC contains this phase. Olivine, in contrast, is a typical phase in mantle rocks and in many silicate rocks, including some of the KVC silicate lithologies. However, the high forsterite content, combined with high Mn (Fig. 12b) and very low Ni contents (Braunger *et al.*, 2018) is atypical of xenocrystic or mantle-derived olivine, but is typically observed in carbonatites (e.g. Lee *et al.*, 2004). At the KVC, olivine and monticellite (both Mn-enriched) are fairly rare and only occur in relatively mica-poor samples. Due to their cracked appearance, occasional rounded edges, serpentinization/iddingitization and partial resorption, a xenocrystic origin cannot be excluded. However, the absence of even small-scale compositional modifications (for example compositional zoning as would be expected by diffusional processes) renders this unlikely. Hence, we suggest that monticellite and olivine may represent silicate minerals of type I, probably representing 'true' primary magmatic silicates (type Ia), although it cannot be completely

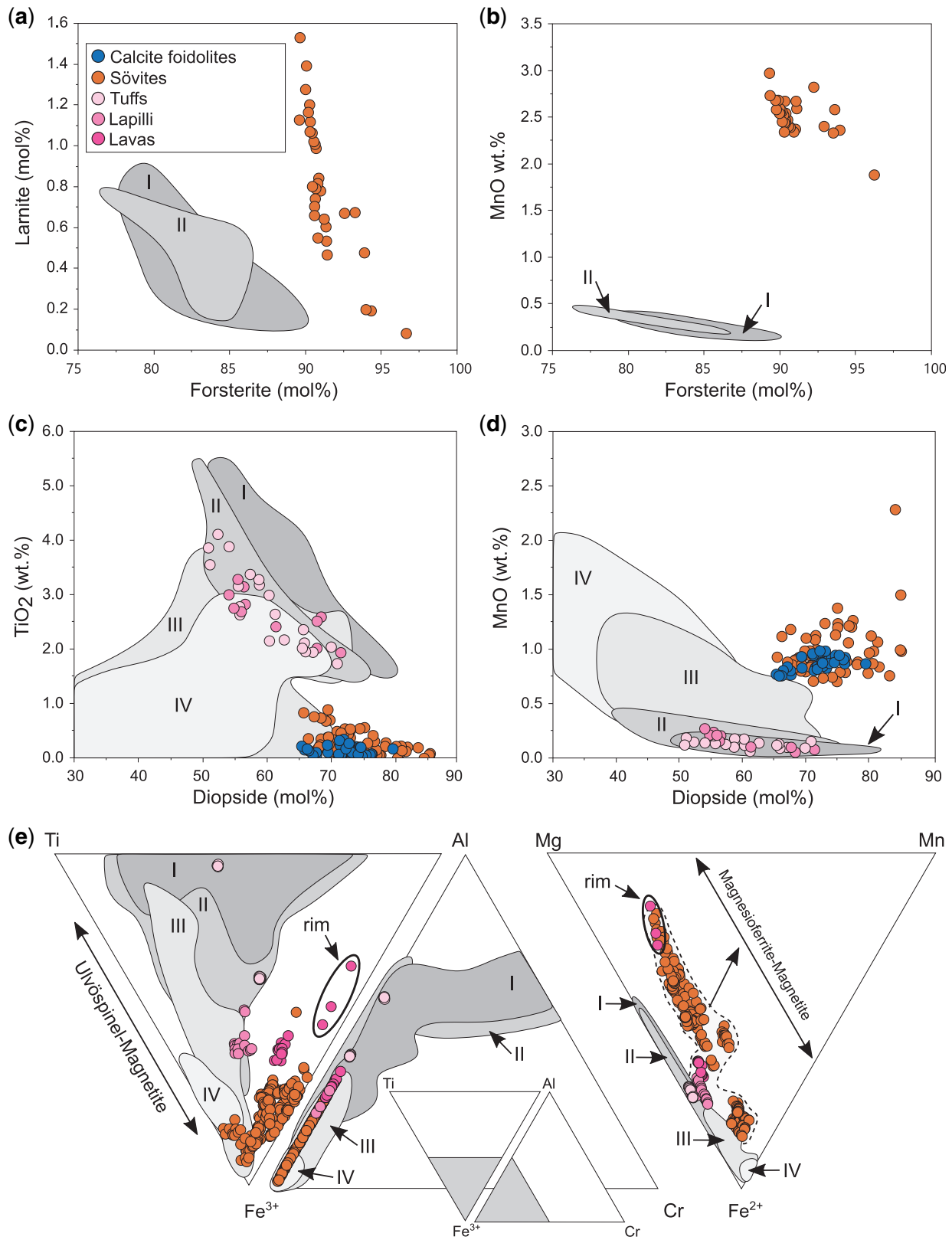


Fig. 12. Composition of olivine, clinopyroxene and spinel group minerals (magnetite-magnesioferrite) from carbonatites and calcite foidolites of the KVC. (a) Larnite vs forsterite components. (b) MnO (wt %) vs forsterite component in olivine. (c) and (d) Concentrations of TiO₂ (wt %) and MnO (wt %) vs diopside component in clinopyroxene. (e) Spinel group mineral compositions from carbonatites of the KVC. Grey fields represent data (olivine, clinopyroxene and spinel group minerals, respectively) for the KVC silicate rocks (Braunger *et al.*, 2018): (I) Olivine nephelinites and limburgites, (II) tephritic rocks, (III) melilititic and häünyitic/häünyolitic rocks, (IV) phonolitic/syenitic rocks, (V) häünyilites and melteigites.

Table 6: Representative EPMA analyses of pyroxene from the Kaiserstuhl Volcanic Complex

Sample #	260	1197	1151	1229	1215	1178	1237	GM1	GM2	GM4
Locality	Badberg							Henkenberg		
Rock type	Calcite foidolite							Crystal tuff		Lapili stone
wt %										
SiO ₂	53.44	49.16	49.40	49.20	49.74	51.84	50.74	44.37	45.04	44.89
TiO ₂	0.05	0.11	0.21	0.58	0.33	0.10	0.17	3.30	2.30	2.64
Al ₂ O ₃	1.60	3.60	3.61	3.58	3.30	1.08	2.81	9.37	8.52	8.38
Cr ₂ O ₃	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.04	0.04	0.03
FeO	6.15	8.81	8.61	7.97	7.25	5.96	6.58	7.53	7.35	7.75
MnO	0.94	0.81	0.83	0.95	0.89	1.14	0.99	0.13	0.15	0.15
MgO	13.93	12.52	12.67	12.71	13.22	14.26	13.65	11.53	11.83	11.51
CaO	24.70	23.90	24.35	23.96	24.23	23.90	24.35	22.77	23.65	23.64
Na ₂ O	0.38	0.47	0.41	0.51	0.40	0.51	0.37	0.53	0.42	0.51
Total	101.19	99.38	100.11	99.47	99.37	98.81	99.67	99.65	99.32	99.56
Formula based on 4 cations and 6 oxygens										
Si	1.96	1.84	1.84	1.84	1.86	1.94	1.88	1.66	1.68	1.68
Ti	0.00	0.003	0.01	0.02	0.01	0.003	0.005	0.09	0.06	0.07
Al	0.07	0.16	0.16	0.16	0.15	0.05	0.12	0.41	0.38	0.37
Cr	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.001	0.001	0.001
Fe ³⁺	0.04	0.18	0.18	0.17	0.15	0.11	0.12	0.12	0.15	0.16
Fe ²⁺	0.15	0.09	0.08	0.08	0.08	0.08	0.08	0.12	0.08	0.09
Mg	0.76	0.70	0.70	0.71	0.74	0.79	0.76	0.64	0.66	0.64
Mn	0.03	0.03	0.03	0.03	0.03	0.04	0.03	0.004	0.005	0.005
Ca	0.97	0.96	0.97	0.96	0.97	0.96	0.97	0.91	0.95	0.95
Na	0.03	0.03	0.03	0.04	0.03	0.04	0.03	0.04	0.03	0.04
Total cations	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00
Mg#	0.84	0.88	0.89	0.89	0.91	0.91	0.90	0.85	0.90	0.88

bdl, below detection limit.

excluded that they are the product of small-scale contamination (type Ib).

Clinopyroxene (cpx) and garnet in extrusive carbonatites are broken and partly rounded (Fig. 5d). As they chemically resemble those of the silicate rocks (Figs 12 and 13), both phases are interpreted as being entrained from the KVC silicate rocks during eruption, representing type II silicates. In sövites, cpx only occurs in the Badberg samples, where it is mostly euhedral and compositionally distinct from those found in extrusive carbonatites and the known silicate rocks of the KVC, but very similar to the recrystallized cpx of calcite foidolites (Fig. 12c and d). Therefore, we suggest that cpx in the Badberg sövites formed similarly to mica (see above) by the interaction between xenoliths and carbonatite magma, representing a type Ib silicate. A possible entrainment of recrystallized cpx from calcite foidolites or an entrainment of original cpx (both representing a type II silicate) from the calcite foidolite precursor lithology (before alteration) is considered unlikely, as cpx in sövite is mostly euhedral (Fig. 3e), whereas it is interstitial and completely recrystallized in calcite foidolites (Fig. 7e). Also, a lack of small-scale compositional modifications (zoning) by xenocryst–carbonatite magma interactions render an entrainment of the original cpx improbable.

Mineralogical and mineral chemical differences among the KVC carbonatites

Based on geophysical and structural data (Brauch *et al.*, 2018; see above) we assume that the various sövite

bodies of the KVC once belonged to a common sub-vertical pipe structure that was intersected, displaced and tilted by subsequent tectonic activity within the Rhine Graben system (Fig. 1c). Consequently, sövites at the Badberg may represent a higher emplacement level (shallower level of the pipe) than the Haselschacher Buck and Degenmatt occurrences. Orberg sövites may represent a large apophysis of this structure (e.g. Hubaux, 1964; Katz-Lehnert, 1989) and based on pyrochlore textures and compositions, these rocks experienced an intense hydrothermal overprint (Walter *et al.*, 2018). Although it is assumed that all four carbonatite localities belong to one batch of melt, major mineralogical and mineral chemical differences between the Badberg sövites and the other carbonatite bodies are prominent:

1. only Badberg sövites contain variable amounts of calcite foidolites (Figs 6 and 7);
2. only Badberg sövites are comparatively rich in mica that is texturally and compositionally distinct from mica in the other sövite occurrences (Figs 4c, d and 6);
3. only some Badberg sövites contain clinopyroxene (Fig. 3e);
4. only apatite in Badberg sövites shows exceptionally Si- and REE-rich rims (Fig. 9a).

In the following we discuss the significance of the calcite foidolites and the possible reasons for the presence of compositionally distinct mica and apatite in the Badberg sövites, including contamination, magmatic differentiation and hydrothermal overprint.

Table 7: Representative EPMA analyses of magnetite from the Kaiserstuhl Volcanic Complex

Sample Locality	1237 Badberg	1136	1215	1151	1362	1410 Orberg III	1413	1353 Orberg IV	237	225 Orberg V	224	1354 Haselschacher Buck	284	1415b Katharinenberg	1415c	1371 Henkenberg	GM2	GM3	GM4	1337 Kirchberg					
Rock type	Sövite																			Crystal tuff			Lapilli stone		Lava
wt %																									
SiO ₂	0.02	0.09	0.12	0.04	0.10	0.04	0.04	0.04	0.02	0.02	0.02	0.02	0.03	0.03	0.12	0.05	0.31	0.05	0.15	0.10	0.04				
TiO ₂	2.81	9.41	3.60	2.07	5.64	1.24	1.91	1.23	1.51	1.29	1.48	1.74	3.04	2.23	2.80	4.04	4.04	3.52	7.57	2.34	1.79				
Fe ₂ O ₃	62.87	46.64	61.37	65.25	56.20	65.74	64.94	67.16	68.68	66.04	65.35	64.87	64.11	64.51	63.38	57.31	61.56	51.17	51.17	62.75	65.90				
FeO	17.87	30.68	26.70	25.84	22.13	18.42	25.93	15.82	15.35	18.69	19.99	26.88	22.06	18.84	20.92	24.08	27.20	28.19	18.06	15.54					
MnO	3.99	0.65	3.08	3.96	3.46	4.62	3.69	4.76	4.85	4.18	3.72	2.89	5.50	4.29	4.31	2.87	2.69	1.74	3.99	5.02					
MgO	8.54	6.73	3.64	2.43	8.08	6.53	2.38	8.18	8.54	6.75	6.36	2.30	4.25	7.25	6.17	6.44	3.36	6.37	8.30	8.55					
V ₂ O ₅	0.41	0.15	0.19	0.25	0.22	0.47	0.18	0.38	0.59	0.59	0.31	0.18	0.91	0.31	0.24	0.08	0.20	0.15	0.41	0.52					
Al ₂ O ₃	5.51	7.39	2.85	1.57	6.36	4.20	1.79	4.07	2.73	4.21	4.45	2.33	1.18	4.15	4.01	6.77	2.59	6.22	6.16	4.35					
Cr ₂ O ₃	0.03	0.04	0.01	0.01	0.03	0.03	0.01	0.02	0.04	0.04	0.02	0.01	0.05	0.02	0.01	0.01	0.01	0.01	0.03	0.03					
ZnO	0.29	0.07	0.13	0.20	0.20	0.28	0.16	0.31	0.28	0.29	0.28	0.19	0.30	0.24	0.30	0.14	0.13	0.12	0.28	0.33					
Total	102.34	101.85	101.67	101.62	102.43	101.58	101.04	101.97	102.59	102.08	102.00	101.43	101.45	101.95	102.20	102.04	101.31	101.67	102.42	102.09					
Formula based on 3 cations and 4 oxygens																									
Si	0.001	0.003	0.004	0.002	0.003	0.001	0.001	0.001	0.000	0.001	0.001	0.001	0.001	0.004	0.002	0.011	0.002	0.005	0.004	0.001					
Ti	0.07	0.24	0.10	0.06	0.14	0.03	0.05	0.03	0.04	0.03	0.04	0.05	0.08	0.06	0.07	0.11	0.09	0.20	0.06	0.05					
Fe ³⁺	1.62	1.21	1.68	1.81	1.45	1.75	1.81	1.76	1.80	1.75	1.73	1.80	1.76	1.70	1.68	1.49	1.69	1.34	1.62	1.72					
Fe ²⁺	0.52	0.88	0.81	0.80	0.63	0.56	0.80	0.47	0.45	0.55	0.60	0.83	0.68	0.56	0.62	0.70	0.83	0.83	0.52	0.45					
Mn	0.12	0.02	0.10	0.12	0.10	0.14	0.12	0.14	0.14	0.12	0.11	0.09	0.17	0.13	0.13	0.08	0.08	0.05	0.12	0.15					
Mg	0.43	0.34	0.19	0.13	0.41	0.34	0.13	0.42	0.44	0.35	0.32	0.13	0.23	0.37	0.32	0.33	0.18	0.32	0.42	0.44					
V	0.01	0.002	0.003	0.004	0.003	0.01	0.003	0.01	0.01	0.01	0.004	0.003	0.01	0.004	0.003	0.001	0.003	0.002	0.01	0.01					
Al	0.22	0.30	0.12	0.07	0.25	0.17	0.08	0.16	0.11	0.17	0.18	0.10	0.05	0.17	0.17	0.28	0.11	0.25	0.25	0.18					
Cr	0.001	0.001	0.00	0.00	0.001	0.001	0.00	0.001	0.001	0.001	0.001	0.00	0.001	0.00	0.00	0.00	0.00	0.00	0.001	0.001					
Zn	0.01	0.002	0.004	0.01	0.01	0.01	0.004	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.004	0.004	0.003	0.001	0.01					
Total cations	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00					

Table 8: Representative EPMA analyses of garnet from the Kaiserstuhl Volcanic Complex

Sample # Location	1158 Badberg	1158	1168	1168	1147	1147	1147	1173	1173	1173	GM1 Henkenberg	GM2 Henkenberg	GM2	GM3	1369	1369	GM4	1337 Kirchberg	1337	
Rock type	Calcite foidolite										Lapilli stone Lava									
wt %	36.36	33.52	36.67	34.21	35.34	27.57	32.00	34.72	32.88	33.62	30.35	33.55	28.38	31.12	32.29	32.83	34.95			
SiO ₂	2.08	3.48	0.82	3.11	2.66	8.60	9.34	4.81	7.02	5.55	10.55	5.45	13.52	9.77	7.27	7.10	3.78			
TiO ₂	0.17	1.12	0.17	2.67	0.44	3.93	0.12	0.08	0.33	0.17	0.38	bdl	0.55	bdl	0.07	0.19	0.32			
ZrO ₂	bdl	bdl	0.07	bdl	bdl	bdl	bdl	bdl	bdl	0.15	0.10	bdl	0.14	bdl	0.17	bdl	bdl			
Y ₂ O ₃	5.55	3.91	4.62	2.63	3.03	1.93	1.41	1.17	2.24	2.66	3.34	2.63	4.20	2.22	3.24	1.66	2.77			
Al ₂ O ₃	22.16	22.82	23.98	23.99	24.87	20.56	25.07	26.84	24.40	25.12	22.24	24.76	20.04	22.13	22.81	25.78	25.84			
Fe ₂ O ₃	0.56	0.55	0.58	0.79	0.62	0.73	0.49	0.50	0.78	0.63	0.52	0.76	0.49	0.62	0.62	0.58	0.94			
MnO	0.69	0.89	0.55	0.89	0.69	1.25	0.46	0.27	0.62	0.49	0.95	0.46	0.95	0.90	0.88	0.45	0.22			
MgO	32.37	33.16	33.11	32.47	31.87	31.59	32.27	32.73	32.50	33.05	32.63	32.19	31.67	31.93	32.82	32.29	32.57			
CaO	0.01	0.02	0.03	0.07	0.04	0.17	0.26	0.21	0.12	0.06	0.08	0.12	0.11	0.09	0.07	0.14	0.08			
Na ₂ O	bdl	bdl	0.04	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.04	0.04	bdl	0.06	bdl	bdl	bdl			
ThO ₂	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
UO ₂	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Ta ₂ O ₅	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Nb ₂ O ₅	0.16	1.57	0.49	1.19	0.45	5.19	bdl	0.11	0.13	0.07	0.14	bdl	0.03	0.22	0.27	bdl	0.09			
SrO	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
La ₂ O ₃	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Ce ₂ O ₃	bdl	bdl	0.10	0.12	bdl	0.10	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.08	bdl	0.07	0.07			
Nd ₂ O ₃	bdl	bdl	bdl	0.09	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.05	0.07			
Pr ₂ O ₃	bdl	0.06	0.05	bdl	bdl	bdl	bdl	bdl	bdl	0.04	bdl	bdl	0.06	bdl	0.07	bdl	0.07			
Sm ₂ O ₃	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.05	0.04	bdl	0.03			
F	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Total	100.11	101.27	101.27	102.23	100.01	101.75	101.43	101.45	101.01	101.62	101.30	100.08	100.13	99.25	100.61	101.13	101.79			
Formula based on 8 cations and 12 oxygens																				
Si	2.98	2.78	2.99	2.84	2.95	2.38	2.67	2.88	2.74	2.78	2.52	2.81	2.39	2.64	2.69	2.74	2.88			
Ti	0.13	0.22	0.05	0.19	0.17	0.56	0.59	0.30	0.44	0.34	0.66	0.34	0.86	0.62	0.45	0.45	0.23			
Zr	0.01	0.05	0.01	0.11	0.02	0.17	0.01	0.00	0.01	0.01	0.02	bdl	0.02	bdl	0.00	0.01	0.01			
Y	bdl	bdl	0.003	bdl	bdl	bdl	bdl	bdl	bdl	0.01	0.004	bdl	0.01	bdl	0.01	bdl	bdl			
Al	0.54	0.38	0.44	0.26	0.30	0.20	0.14	0.11	0.22	0.26	0.33	0.26	0.42	0.22	0.32	0.16	0.27			
Fe ²⁺	0.16	0.06	0.08	0.16	0.18	0.32	0.20	0.15	0.13	0.08	0.12	0.12	0.23	0.20	0.06	0.17	0.14			
Fe ³⁺	1.21	1.36	1.40	1.34	1.39	1.01	1.38	1.53	1.40	1.48	1.27	1.44	1.05	1.22	1.36	1.45	1.46			
Mn	0.04	0.04	0.04	0.06	0.04	0.05	0.03	0.04	0.05	0.04	0.04	0.05	0.04	0.04	0.04	0.04	0.07			
Mg	0.08	0.11	0.07	0.11	0.09	0.16	0.06	0.03	0.08	0.06	0.12	0.06	0.12	0.11	0.11	0.06	0.03			
Ca	2.85	2.94	2.90	2.88	2.85	2.92	2.89	2.91	2.90	2.93	2.91	2.89	2.86	2.91	2.93	2.89	2.88			
Na	0.002	0.00	0.00	0.01	0.01	0.03	0.04	0.03	0.02	0.01	0.01	0.02	0.02	0.01	0.01	0.02	0.01			
Th	bdl	bdl	0.00	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.00	0.00	bdl	0.00	bdl	bdl	bdl			
U	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Ta	bdl	0.004	bdl	bdl	bdl	0.003	bdl	bdl	bdl	0.001	bdl	0.003	0.001	0.001	bdl	bdl	0.003			
Nb	0.01	0.06	0.02	0.04	0.02	0.20	bdl	0.004	0.01	bdl	0.01	bdl	bdl	0.01	0.01	bdl	0.003			
Sr	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
REE total	bdl	bdl	0.004	0.006	bdl	0.003	bdl	bdl	bdl	bdl	bdl	bdl	0.002	0.005	0.003	0.004	0.007			
F	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl			
Total cations	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00	8.00			

bdl, below detection limit. Noticeable variation of representative analyses in a single sample has been indicated by the addition of further analyses, reflecting the variation.

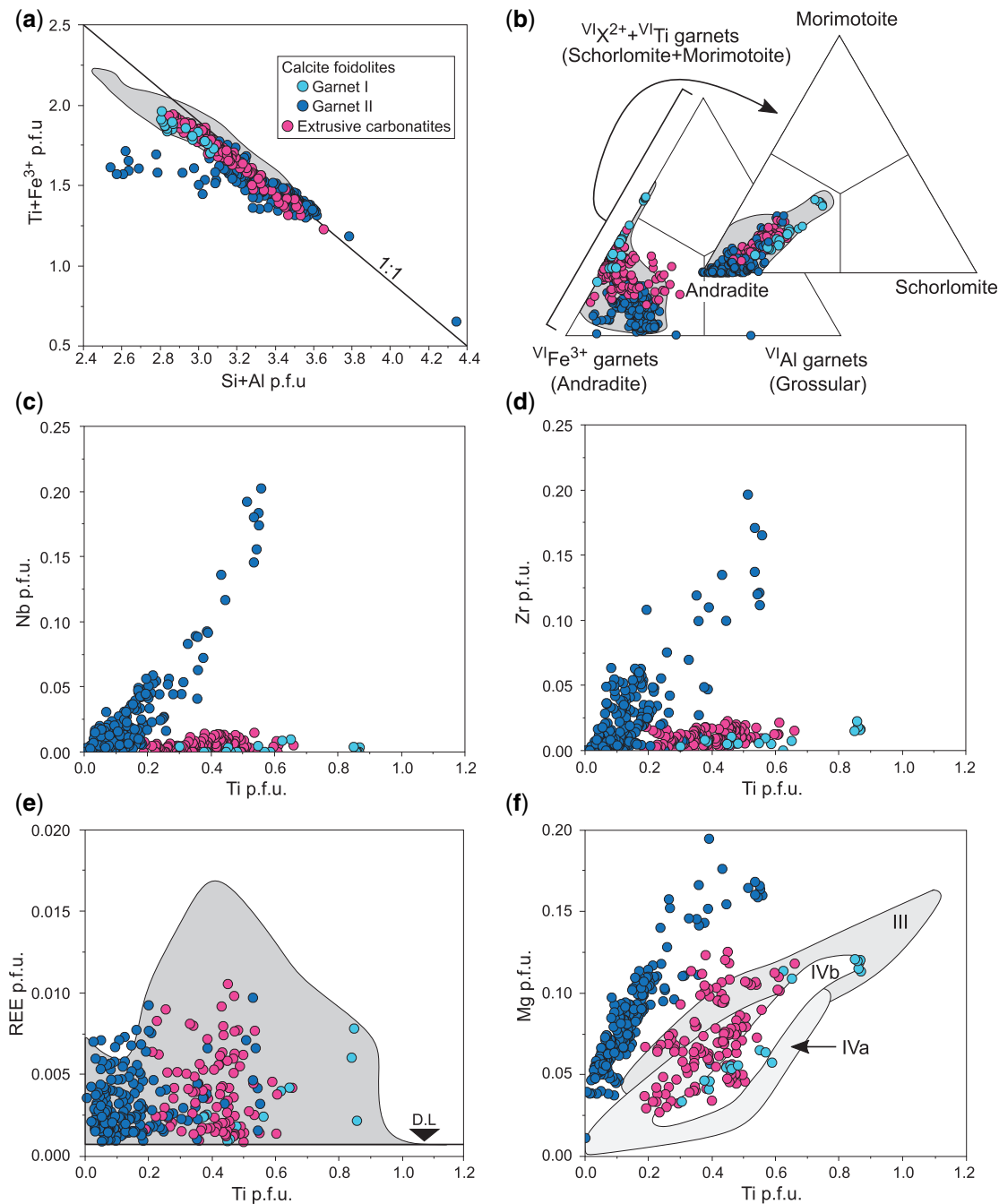


Fig. 13. Compositional variation of garnet from carbonatites and calcite foidolites of the KVC. (a) $Ti+Fe^{3+}$ vs $Si+Al$. The solid line illustrates the ideal 1:1 substitution. (b) Garnet composition based on the octahedral-site occupations Fe^{3+} , $X^{2+}+Ti$ ($X= Mg, Fe^{2+}$) and Al and based on the endmembers andradite, morimotoite and schorlomite. (c)–(f) Concentrations of Nb, Zr, REE and Mg vs Ti, with data from the KVC silicate rocks shown as grey fields for comparison (Braunger *et al.*, 2018): (III) melilititic and haüynitic/haüynolitic rocks, melteigites, (IVa) nasean syenites (former ‘ledmorites’), (IVb) phonolites and nasean syenite inclusions therein.

Significance of calcite foidolites (carbonatite magma–xenolith interaction)

Calcite foidolites in the Badberg sövites (Figs 6 and 7) probably do not represent cumulates of the carbonatite magma itself, since their mineralogy is very different from the sövites and large differences in the modal abundances would be expected. Also, feldspar, which is found as relicts in the calcite foidolites, is generally unstable at the low silica activities expected for

carbonatite magmas (Barker, 2001; Massuyeau *et al.*, 2015). Therefore, we assume that these rocks represent equivalents of previously emplaced silicate rocks entrained in and subsequently resorbed and metasomatized by the intruding carbonatitic magma.

Despite their textural resemblance to haüynolites that occur as xenoliths in some phonolitic rocks (Czygan, 1977), the chemical composition of relict garnet (garnet type I) in calcite foidolites suggests that the

Table 9: Representative EPMA analyses of zirconolite from the Kaiserstuhl Volcanic Complex

Sample #	1354	1354	C 71	C 72
Locality	Haselschacher Buck		Kaiserstuhl	
Rock type	Sövite			
wt %				
CaO	12.56	12.52	11.38	12.20
Na ₂ O	bdl	bdl	n.a.	n.a.
SrO	0.29	bdl	n.a.	n.a.
La ₂ O ₃	0.16	0.30	n.a.	n.a.
Ce ₂ O ₃	1.71	2.02	0.77	0.90
Pr ₂ O ₃	0.37	0.34	n.a.	n.a.
Nd ₂ O ₃	1.03	1.13	1.10	n.a.
Sm ₂ O ₃	0.27	0.19	0.27	n.a.
Y ₂ O ₃	0.48	0.49	n.a.	n.a.
ZrO ₂	33.75	32.68	30.51	34.80
UO ₂	1.04	1.23	1.22	1.40
ThO ₂	1.18	1.65	5.13	4.10
TiO ₂	10.81	10.66	13.56	22.70
Nb ₂ O ₅	21.20	21.53	22.07	15.70
Ta ₂ O ₅	4.10	4.45	3.08	n.a.
MgO	n.a.	n.a.	0.85	n.a.
Al ₂ O ₃	0.91	0.82	n.a.	n.a.
FeO	8.00	7.77	7.41	7.60
MnO	0.98	1.03	0.94	0.20
SiO ₂	bdl	bdl	n.a.	n.a.
F	bdl	bdl	n.a.	n.a.
Total	98.83	98.80	98.29	99.60
Formula based on 4 cations				
Ca	0.90	0.91	0.84	0.87
Na	bdl	bdl	n.a.	n.a.
Sr	0.01	bdl	n.a.	n.a.
REE+Y	0.10	0.12	0.05	0.02
Zr	1.11	1.07	1.02	1.08
U	0.01	0.02	0.02	0.02
Th	0.02	0.03	0.08	0.06
Ti	0.55	0.54	0.70	1.10
Nb	0.64	0.66	0.67	0.46
Ta	0.07	0.08	0.06	n.a.
Mg	n.a.	n.a.	0.09	n.a.
Al	0.07	0.07	n.a.	n.a.
Fe	0.45	0.44	0.42	0.38
Mn	0.06	0.06	0.05	0.01
Si	bdl	bdl	n.a.	n.a.
F	bdl	bdl	n.a.	n.a.
Total cations	4.00	4.00	4.00	4.00

bdl, below detection limit. n.a., not analysed.

C71: Data from Keller (1984). C72: Data from Sinclair & Eggleton (1982).

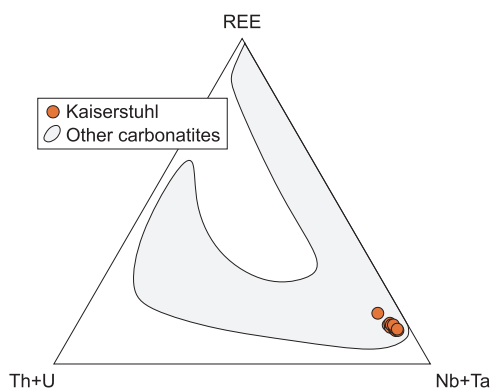
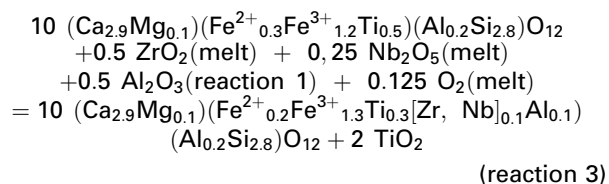
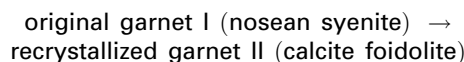
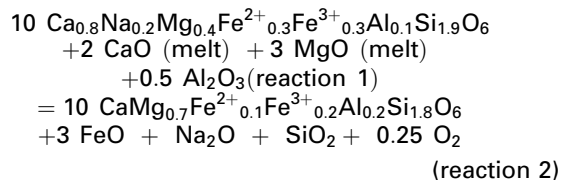
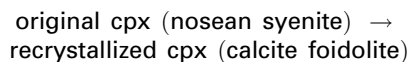
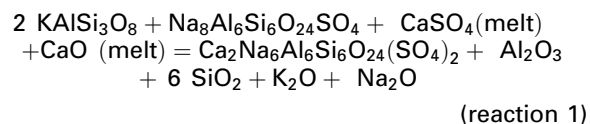
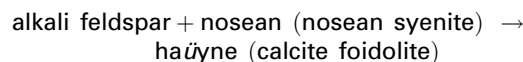


Fig. 14. Compositional variation of zirconolite from the KVC in comparison with other carbonatites (data from Williams & Gieré, 1996; Hurai *et al.*, 2018).

nosean syenites that underlie the Badberg body (as evidenced by the FB drill core; see Walter *et al.*, 2018), are the most likely protolith (Fig. 13). However, in contrast to nosean syenites, calcite foidolites have a higher proportion of sodalite-group minerals, alkali feldspar is almost absent and clinopyroxene and garnet are subhedral to anhedral. We attribute these mineralogical and textural differences to the interaction with the carbonatite magma (Fig. 16).

Carbonatites reach liquidus temperatures above 1000 °C (e.g. Kharlomov, 1981; Panina, 2005; Guzmics *et al.*, 2011) and contain large amounts of volatile phases (e.g. F and H₂O; Treiman & Essene, 1984; Jago & Gittins, 1991). Therefore, they can cause intense metasomatism (Elliot *et al.*, 2018) and may be capable of partly assimilating silicate rocks. Textural evidence for the interaction between the entrained xenoliths and the carbonatite magma includes: (i) the almost complete replacement of precursor alkali feldspar by foids; (ii) the compositional change of nosean to haüyne; (iii) blackwall-like mica seams around the xenoliths; (iv) the occurrence of recrystallized garnet (garnet type II) with relics of original garnet cores (garnet type I); and (v) crosscutting carbonatitic veins (Figs 6 and 7). The modifications in nosean syenites can be expressed by the schematic mineral reactions (1) – (3):



Garnet II has relatively high Nb and Zr contents (Fig. 13), reflecting the importance of the carbonatitic magma during metasomatism, as carbonatites are typically enriched in these elements (usually stored in pyrochlore, perovskite, baddeleyite and zirconolite). Similar modifications of silicate rocks by carbonatitic melts have been described as ‘antiskarn’

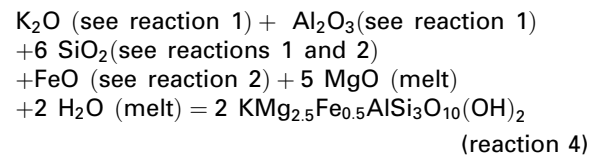
metasomatism (Anenburg & Mavrogenes, 2018) and reflect the mutual interaction during contamination.

Importantly, xenoliths in the lower sections of the KB3 drill hole, which represents the central part of the tilted Badberg body, are much more strongly disaggregated, which may indicate higher degrees of resorption compared to the marginal parts of the body. This may be attributed to a longer lasting heat flow that predominantly occurred in the central area of the carbonatite pipe, while marginal zones experienced an earlier and more rapid cooling. Consequently, this central zone contains more silica that may have induced the observed exceptional mica enrichment in those parts (Fig. 6). In these samples (down-hole below +100 metres above sea level), pyrochlore is enriched in Fe^{2+} , Al and Ti (Walter *et al.*, 2018), which was probably released by the resorption of xenoliths. Note that such enrichment of Fe^{2+} and Al in pyrochlore is confined to the Badberg locality and does not exist at the other carbonatite localities of the KVC.

The significance of mica composition

The enhanced formation of mica (and the formation of diopside instead of monticellite/olivine) in the Badberg samples cannot be ascribed to magmatic differentiation, since silicates in carbonatites crystallize early, which would decrease the SiO_2 concentration and silica activity in the carbonatite magma with fractionation (e.g. Krasnova *et al.*, 2004; Lee *et al.*, 2004). However, strong textural evidence and mineral compositions (see above) suggest a connection between the formation of mica (and cpx) and the entrainment of xenoliths in the sövites from Badberg.

In contrast to mica from the other sövite bodies of the KVC (which are dominated by the kinoshitalite substitution), mica at Badberg is dominated by the eastonite substitution (see above). This indicates that the availability of Al exceeds the availability of Ba, most likely due to the provision of Al by the resorption of feldspar (reaction 1). Furthermore, Badberg mica shows an unusual enrichment in octahedral Fe^{2+} . This is atypical for the other sövites of the KVC and carbonatites in general (e.g. Kovdor, Palabora, Sokli and Jacupiranga; Fig. 15), where the formation of tetraferriphlogopite (incorporation of Fe^{3+}) or Fe-poor phlogopite is preferred over the formation of Fe^{2+} -rich mica (biotite; e.g. Lee *et al.*, 2003; Krasnova *et al.*, 2004; Giebel *et al.*, 2019). While all other localities of the KVC show 'typical carbonatite micas' with respect to their Mg, Al and Fe concentrations, Badberg micas are more similar to micas from silicate rocks (Fig. 15), which we ascribe to the metasomatic interaction with the calcite foidolite clasts and rafts (see above). Alteration reactions (1) – (3) above demonstrate the release of K, Al, Si and Fe^{2+} , which are consumed to form Fe^{2+} -bearing mica (Figs 4c and 7a) according to the schematic reaction:



Besides the Fe^{2+} -enrichment, the micas of the Badberg contain elevated amounts of Ti and Mn. While Mn and also F are preferably incorporated into the mica precipitating directly on the resorbed clasts (black-wall

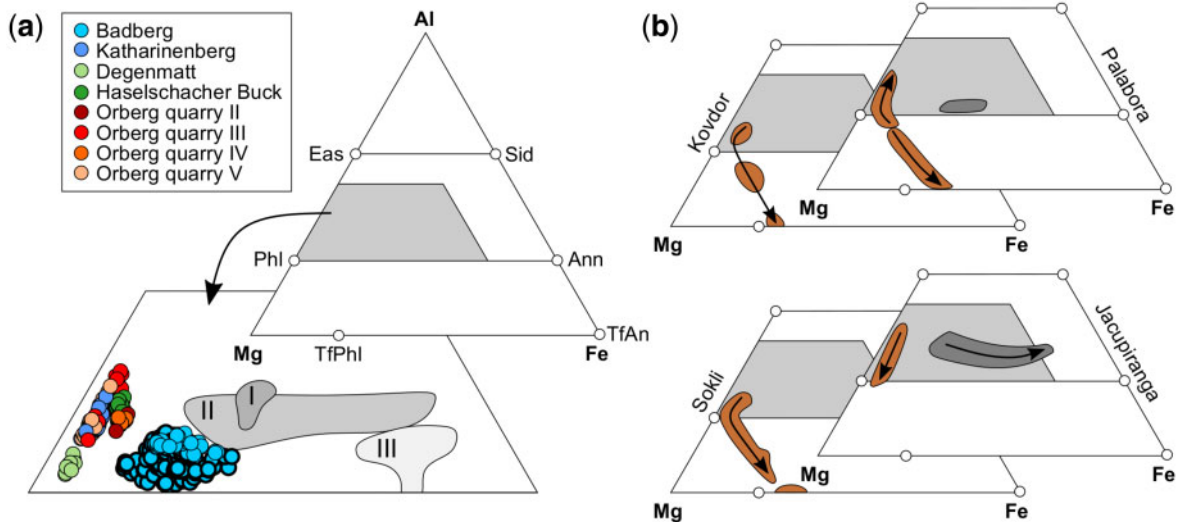


Fig. 15. (a) Ternary diagrams of Mg–Al_{total}–Fe_{total} reflecting the compositional characteristics of mica from the sövites of the KVC, with numbered grey areas representing mica compositions from the silicate rocks of the KVC (Braunger *et al.*, 2018): (I) olivine nephelinites; (II) tephritic rocks; (III) haüyne melilitites. (b) Mica compositions from other carbonatites with brown fields indicating mica compositions in carbonatites and dark grey field mica compositions from associated silicate rocks (Palabora, Kovdor, Sokli and Jacupiranga; Brod *et al.*, 2001; Lee *et al.*, 2003; Krasnova *et al.*, 2004; Giebel *et al.*, 2019). Ann, annite; Eas, eastonite; Sid, siderophyllite; Phl, phlogopite; TfPhl, tetraferriphlogopite; TfAn, tetraferriannite.

mica), Ti, Na and Ba behave mostly conversely and appear to have been enriched in the carbonate melt, further distant from the resorbed clasts (Fig. 11). The enhanced crystallization of mica at Badberg and the related consumption of Mg from the carbonatite melt may also be the reason for the relative depletion of Mg in subsequently crystallized calcite (Fig. 8), due to the generally very rapid diffusion in carbonatite melts and their resulting very fast equilibration (e.g. Anenburg *et al.*, 2018 and references therein). Besides Mg, Ba was also derived from the carbonatite melt itself, while the enrichment of Ti (reaction 3) and additional Mn is probably related to the resorption and recrystallization of clinopyroxene and garnet I. The absence of Fe²⁺-poor mica cores suggests that the entire mica crystallization in the Badberg sövites took place during nosean syenite-carbonatite magma interaction. Hence, mica composition cannot be easily used to track magmatic differentiation at Badberg, but may still be applicable at the other localities.

Enrichment of Ba in carbonatites is generally ascribed to differentiation because of its incompatible character (e.g. Kogarko *et al.*, 2012; McCormick & Heathcote, 1987; McCormick & Le Bas, 1996). Moreover, Ba is efficiently mobilized in carbonatites, is strongly concentrated in residual liquids and dissolves favourably in aqueous fluids (Anenburg & Mavrogenes, 2018). Thus, the strong enrichment of Ba in the mica from the Orberg sövites (Fig. 11) confirms the assumption of Walter *et al.* (2018) that the Orberg sövites are evolved and hydrothermally overprinted. The high Mg# of Orberg mica is attributed to the relative enrichment of Mg over Fe during differentiation (fractional crystallization) due to the contemporaneous formation of Fe-rich spinel group minerals. This is the reverse of what is typically expected during silicate-magma differentiation (dependent on the redox state of the magma; e.g. Heathcote & McCormick, 1989; Giebel *et al.*, 2019). A comparison between the enrichment of Ba and the increase of the Mg# in these micas suggests a much stronger sensitivity of the former in terms of mica evolution. In conclusion, mica does not only record differentiation, but also contamination processes in carbonatites exceptionally well.

The significance of apatite composition

The local increase in silica activity in the carbonatite melt (reactions 1 & 2) enhanced the incorporation of Si in apatite (Figs 9 and 10) and favoured simultaneous REE incorporation due to the britholite substitution (e.g. Hammouda *et al.*, 2010). In contrast to cores of Badberg apatites, which generally reveal a lower britholite substitution, rims are significantly more strongly affected (Fig. 9a). The sharp contact between the cores and rims suggests an abrupt and strong compositional change of the carbonatite magma during apatite formation, which excludes continuous magma differentiation, although

the generally high Sr contents in Badberg apatites (Fig. 9c) indicate a relatively evolved character for the Badberg sövites.

In contrast to a few apatites from the Orberg that occur in strongly hydrothermally altered samples and experienced a belovite substitution (see above; Fig. 9), which is typical for hydrothermal apatite evolution in many carbonatites (e.g. de Toledo *et al.*, 2004; Doroshkevich *et al.*, 2009), a hydrothermal origin of the Badberg apatites in respect of the increased britholite substitution (0.3–0.5 REE+Si pfu) in its rims is also unlikely. This phenomenon is restricted to magmatic temperatures (>600 °C), while hydrothermal conditions (including sufficient SiO₂) promote the formation of REE-poorer apatite and REE-rich silicate minerals (e.g. cerite; Anenburg & Mavrogenes, 2018; Anenburg *et al.*, 2018). In addition, typical hydrothermal interaction textures (reprecipitation/replacement reactions; e.g. Giebel *et al.*, 2017) are typically absent in Badberg sövites.

We assume that the enhanced britholite substitution in apatites is related to the interaction between the carbonatite magma and silicate xenoliths. The core-rim variation records the time-dependent effect of contamination. This is additionally reflected by the increasing Si and REE content in apatite in the deeper sections of drill core KB3 (Fig. 10; central zones of the pipe), where a stronger disaggregation of xenoliths caused an increased release of silica (see above). Experimental data (Klemme & Dalpé, 2003) confirm enhanced partitioning of REE into apatite with increasing SiO₂ content in the melt. This emphasizes the potential importance of contamination of carbonatitic magmas for REE mineralization; the local enrichment of Si favours an early (magmatic) incorporation of REE into apatite, which inhibits the subsequent enrichment of REE in residual liquids/fluids. These late-stage fluids, however, are the most common source for the deposit-quality enrichment of REE-minerals and/or REE-rich minerals in carbonatites (Wall & Mariano, 1996). This connection is convincingly indicated by a comparison of the REE concentrations in apatite and pyrochlore between the Badberg and other carbonatite bodies (Fig. 17). While Badberg apatites are REE-rich (due to contamination; Fig. 9a), the incorporation of REE in Badberg (late-stage) pyrochlore seems to be largely unaffected. In contrast, apatites at other localities are relatively REE-poor, while hydrothermally overprinted pyrochlore at these localities is strongly REE-enriched (see details in Walter *et al.*, 2018). This potentially has economic implications, although it should be noted that calcite (due to its high modal content) further contributes to the REE budget of the rocks.

Mica in carbonatites – a potential clue to wall rock interaction processes?

Many carbonatites contain appreciable amounts of mica (e.g. Brod *et al.*, 2001; Lee *et al.*, 2003; Reguir *et al.*,

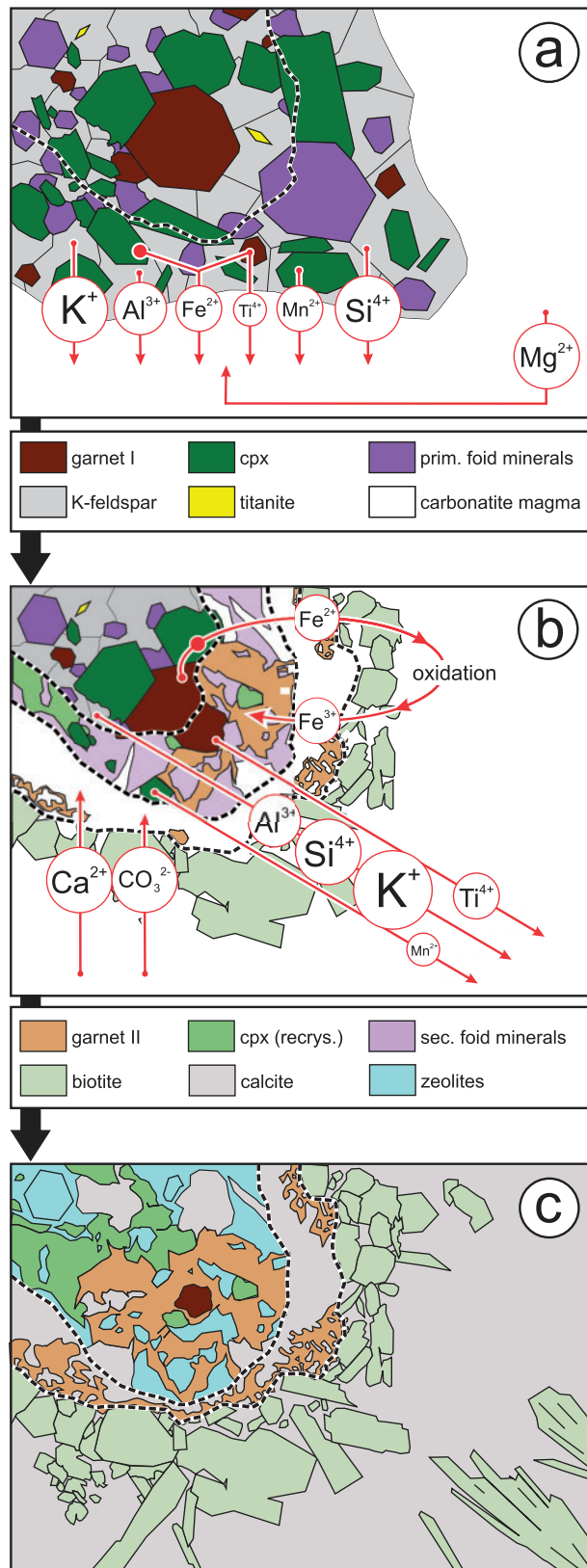


Fig. 16. Schematic diagram illustrating processes during contamination of carbonatitic melt by wall rock. (a) Interaction of carbonatitic melt with xenoliths of nosean syenite leads to resorption of primary minerals in marginal zones and associated release of some elements into the melt. (b) The released elements cause the formation of mica in black-walls, with Mg

2009; Giebel *et al.*, 2019). In many carbonatites mica is more or less equally distributed in the rock, but some carbonatites contain cm- to m-sized mica-rich areas as schlieren and lenses and even mica-dominated rocks (glimmerites) occur in some cases at the contact between carbonatites and the host rocks (glimmeritization; e.g. Gittins *et al.*, 1975; Hoatson *et al.*, 2011).

Our study shows that interaction of carbonatite magma with silicate rocks may induce mica crystallization in carbonatites due to the mobilization and redistribution of K, Si, Al, Mg, Fe and H_2O (reaction 4). Similarly, Vuorinen & Skelton (2004) and Chakhmouradian *et al.* (2008) showed that mica with an unusual Fe^{2+} -rich composition in the Alnö and Eden Lake sövites most probably is the product of the reaction between the carbonatite melt and the silicate host rocks. Thereby, the general question arises whether 'pure' carbonatites are capable of crystallizing mica in larger quantities at all, or whether an additional enrichment of K, Al, Si and Fe by contamination and interaction with silicate wall rocks is critically important. Sufficient Mg occurs in the carbonatite magma itself and experimental studies have shown that carbonate melts are capable of containing relatively high concentrations of H_2O (e.g. 10 wt % at 1 kbar; Keppler, 2003). However, the available experimental data indicate that carbonatites can dissolve only minor amounts of Al (usually <1 wt % Al_2O_3 ; Brooker & Kjarsgaard, 2011) and Si (<2.9 wt % SiO_2) at conditions pertinent to the KVC (750–1200 °C, subvolcanic pressures; Weidendorfer *et al.*, 2017).

We performed mass balance calculations based on these experimental constraints, typical Al and Si concentrations in mica from the Badberg (and various other carbonatites) and carbonatite melt densities (2.2–2.6 g/cm³) estimated for middle to shallow crustal levels at 800–1200 °C (e.g. Wolff, 1994; Genge *et al.*, 1995; Dobson *et al.*, 1996; Kono *et al.*, 2014). These calculations indicate that only minor modal amounts of mica (<7%) can form from a carbonatitic melt – assuming that sufficient K, Mg and H_2O are available and no other silicate minerals are present (ES1, Supplementary Data; supplementary data are available for downloading at <http://www.petrology.oxfordjournals.org>). Consequently, larger amounts of mica in carbonatites would require an additional external introduction of Si and Al. In fact, the enrichment of Al in some carbonatites has been ascribed to the resorption of feldspar from fenitized wall rocks (McCormick & Le Bas, 1996), which obviously also releases alkalis and Si. Further, because of the relatively early fractionation of mica in

Fig. 16. Continued mostly provided by the carbonatitic melt. Contemporaneously Fe^{3+} , Ca and CO_3^{2-} are incorporated in recrystallized garnet II, diopsidic clinopyroxene II and calcite. (c) Completely resorbed/re-crystallized xenolith (now represented as calcite foidolite) with relict garnet I; primary nosean is completely replaced by zeolite.

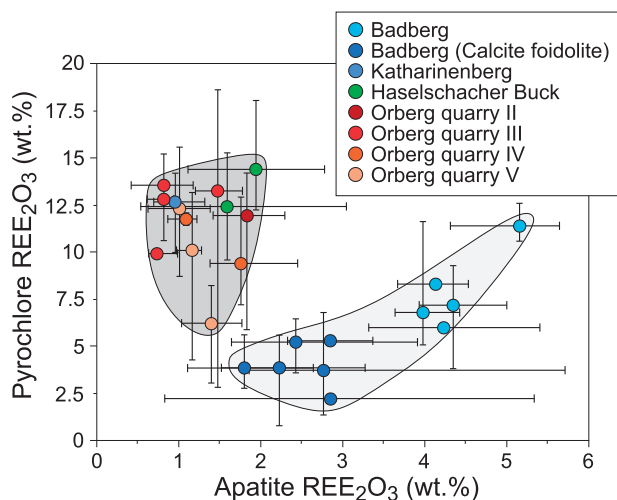


Fig. 17. Comparison of REE concentrations in pyrochlore and apatite in sövites from different localities in the KVC. The error bars represent the maximum or the minimum values, respectively, for pyrochlore/apatite within a sample.

carbonatites, rather mica-poor carbonatites are generally expected at shallow crustal levels. This further excludes the possibility that a compositionally distinct carbonatite magma is responsible for the unusual enrichment of mica in the Badberg sövites. Hence, contamination may play a key-role for enhanced mica formation in evolved carbonatitic systems, such as the KVC. This notion is most likely also applicable for the formation of other silicate minerals (e.g. Anenburg & Mavrogenes, 2018). Here, though, different processes have to be distinguished, including: (1) assimilation of wall rocks/crystals (occurrence of xenoliths/xenocrysts); (2) interaction with wall rocks via diffusion along compositional gradients (wall rock modifications); and (3) introduction of alkali-rich fluids (not easily detectable).

CONCLUSIONS AND IMPLICATIONS

It is generally supposed that crustal contamination plays a negligible role during carbonatite intrusion due to the high rate of magma ascent, extremely low viscosity and occasionally extraordinary low temperatures of carbonatitic magmas (Treiman & Schedl, 1983; Jones *et al.*, 2013 and references therein). At the same time, carbonatitic magmas are known to be extremely reactive because of their massive compositional contrast to almost all crustal rock types and cogenetic silicate rocks (the strong geochemical gradients result in intense diffusional processes) and because of their high amounts of fluxing agents such as P, F and other volatiles, causing various types of metasomatic reactions in deep-seated mantle regions as well as at shallow crustal levels.

Especially shallow-level carbonatites in active rift zones, of which the Kaiserstuhl Volcanic Complex (Southwest Germany) is a good example, may be associated with brittle deformation during and after magma

emplacement, which may support disaggregation and entrainment of fragments and rafts of country rocks and/or cogenetic intrusive rocks. Sub-surface levels may even promote the fragmentation of wall rocks by explosive eruptions, where fragments of wall rock may be partly resorbed by carbonatite melt prior to consolidation.

Our study presents textural and geochemical evidence that intense interaction between carbonatite magma and fragments of earlier, cogenetic, intrusive silicate rocks at a shallow emplacement level can indeed influence the mineralogy and mineral chemistry of sövites. This contamination process causes a mineralogical and geochemical modification of both carbonatite and silicate rocks.

Mica that crystallized from such locally contaminated carbonatites may incorporate relatively high amounts of octahedral Fe^{2+} , which is atypical of carbonatites in general. Such contamination processes may even influence the timing, type and intensity of REE mineralization and hence may be able to cause a relocation of economic levels in a carbonatitic system. In the Badberg case, wall rock interaction causes preferred incorporation of REE in apatite during the magmatic stage. This resulted in relative REE depletion during the late-magmatic/hydrothermal stages. At Orberg, however, contamination processes probably did not play an important role and indeed, REE enrichment in pyrochlore in these samples is much more intense during the hydrothermal stage. However, the otherwise commonly observed formation of late magmatic and hydrothermal REE-F-carbonate minerals (e.g. bastnäsite, synchysite, parisite) in carbonatites is largely lacking at the KVC.

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SUPPLEMENTARY DATA

Supplementary data are available at *Journal of Petrology* online.

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Appendix VI

Submitted manuscript

Study 6

Braunger, S., Scharrer, M., Marks, M. A. W., Wenzel, T., Markl, G., submitted. Hydrothermal halite precipitation in mafic volcanic rocks: the case of the Buggingen salt deposit, SW Germany. *Canadian Mineralogist*.

23 salt deposits and potential liquid mingling and/or assimilation processes. Halite formation in
24 the dyke rocks is rather related to a later, post-magmatic hydrothermal fluid that previously
25 interacted with the salt-rich host rocks. Alteration of the initially glassy groundmass to smectites
26 and zeolites caused an enrichment of Na in the residual fluid, but halite saturation was not
27 reached, as recorded by the absence of groundmass halite. Only fluid rock interaction in mm-
28 sized vugs caused halite-precipitation via desiccation by swelling of previously formed clay
29 minerals. Locally, datolite formed in pseudomorphs after olivine. Its precipitation was
30 controlled by Si and B supply provided by the breakdown of serpentine and smectite.

31

32 *Keywords: potash salt deposit, clay minerals, hydration, olivine melilitite, monchiquite*

33

34

INTRODUCTION

35 Little is known about the interaction between silicate magmas and salt rocks, where the
36 large chemical contrast provides potential for intense magmatic and hydrothermal processes.
37 Studies on basaltic rocks that intruded into the Zechstein evaporates (Upper Permian) of the
38 Werra-Fulda region (Central Germany) report peperite textures and glassy halite inclusions that
39 were interpreted to indicate carnallite-magma mingling (Schofield *et al.*, 2014) and partial
40 melting of salts (Koch & Vogel, 1980). Hydrothermal redistribution of the salt is demonstrated
41 by halite-bearing amygdules and altered ocelli (e.g. Koch, 1978, Knipping, 1989). Besides the
42 Werra-Fulda region, primitive dyke rocks that contain halite-bearing amygdules/ocelli were
43 also reported from the potash salt deposit at Buggingen in southwest Germany (e.g.
44 Wimmenauer, 1951, Braitsch *et al.*, 1964, Hurre, 1973, Esslinger, 1976, Hurre, 1976). Despite
45 a detailed petrographic investigation of the dyke rocks by Hurre (1973), the precipitation
46 mechanism for such salt textures was not investigated so far. Especially intriguing and not

47 understood is the fact that salt crystals do not occur in the groundmass but are restricted to
48 amygdules/ocelli. We shed light on this peculiarity based on a detailed textural, mineralogical
49 and mineral chemical characterization of the monchiquitic and olivine melilititic dyke rocks at
50 the Buggingen salt deposit.

51

52

GEOLOGY

53 The 4.5 m thick potash salt deposit near Buggingen is situated in the Upper Rhine
54 Graben, about 25 km southwest of Freiburg (Fig. 1a), and was mined from 1928 to 1973 at a
55 depth level of 800 to 1000 m (Esslinger, 1976). The deposit is of Rupelian age (34 to 28 Ma;
56 Braitsch *et al.*, 1964, Sissingh, 1998), and is part of the Upper Saliferous Zone that consists of
57 sylvite, halite, anhydrite and marl. It is underlain by brackish-marine calcite- and dolomite-
58 bearing marls of the Middle Saliferous Zone (Zone fossilière). The hanging wall comprises
59 evaporate-rich marls bearing anhydrite, gypsum and halite (Fig. 1b). A progression of the
60 Paratethys and a subsequent regression is indicated by the uppermost layers which consist of
61 bituminous marine shales (Fish Shale), marine-brackish (Meletta Beds) and limnic marls
62 (Cyrena Marls). Unconformable Quaternary sediments (sands and gravel) cover the sequence.
63 Magmatism of the Upper Rhine Graben and outside to the east at Urach and Hegau is part of
64 the European Cenozoic Volcanic Province (Fig. 1a; Wilson & Downes, 1991). The famous
65 Kaiserstuhl volcano (e.g. Braunger *et al.*, 2018, Walter *et al.*, 2018, Giebel *et al.*, 2019) is only
66 20 km away. The first dyke rocks at Buggingen were discovered during mining in 1936 and
67 intruded along two fault zones: The dyke system consists of several olivine melilitites, with
68 additional monchiquites in its western part. A rough estimate of the dyke rock age is given by
69 the nearby, mineralogically similar olivine nephelinites at Eichstetten, Lehen and Mahlberg,
70 which indicate an age of ~13-19 Ma (based on whole rock K-Ar, and U-Pb in perovskite;
71 Baranyi *et al.*, 1976; Binder, personal communication).

72

SAMPLE LOCALITIES

73 The investigated samples were provided by the Regierungspräsidium Freiburg,
74 Landesamt für Geologie, Rohstoffe und Bergbau (Geological Survey of Baden-Württemberg)
75 and were collected in the 1950's and 1960's. Samples from the western dyke zone were taken
76 from the 793 m working horizon within the potash salt deposit and comprise olivine melilitites
77 and monchiquites. At the eastern dyke zone, olivine melilitites in contact to the Fish Shale Zone
78 were sampled in the shaft Heitersheim at a depth of 185-190 meters (Fig. 1b).

79

80

METHODS

81 The investigated thin sections were prepared and polished completely dry (water-free)
82 in order to prevent dissolution and redistribution of halite and sylvite. Mineral compositions
83 were analyzed using a JEOL JXA8900 electron microprobe in wave-length dispersive mode at
84 the Department of Geosciences, University of Tübingen (Germany) with an acceleration
85 voltage of 20 kV and a beam current of 20 nA. Peak counting times for major elements were
86 16 s and between 30 and 60 s for minor elements while the background counting times were
87 half as long. Natural and synthetic standards were used for calibration. Peak overlap corrections
88 involving Ba-Ti, Ti-Ba, V-Ti, V-Cr, Al-Ba, F-Fe, Sm-Ce and Pr-La were implemented and the
89 internal ZAF (for spinel group minerals, garnet, perovskite and titanite) or $\phi\rho z$ correction was
90 applied for the raw data (Armstrong, 1991). Measurements were done with a focused beam
91 except for amphibole and biotite which were measured with a 2 μm spot.

92

93

MINERALOGY AND PETROGRAPHY

94 According to the guidelines of the International Union of Geological Sciences,
95 Subcommittee on the Systematics of Igneous Rocks (Le Maitre *et al.*, 2002), the two types of

96 dyke rocks at Buggingen are monchiquites and olivine melilitites (formerly also described as
97 ankaratrites, a biotite-bearing variety of olivine nephelinites). In the following, the general
98 mineral assemblages of the two dyke rock types as well as the mineralogical differences
99 between dyke rocks associated with the Fish Shale and the potash salt layer are described in
100 more detail. Paragenetic schemes summarize the crystallization sequence for the olivine
101 melilitites and the monchiquites (Fig. 2) and representative mineral compositions for olivine,
102 clinopyroxene, spinel group minerals, amphibole, biotite, garnet, titanite and perovskite are
103 given (Tables 1-4).

104

105 *General mineral assemblages*

106 *Olivine melilitite* Precursor olivine macrocrysts of up to 1 mm in size are
107 completely altered (Fig. 3a) and replaced by serpentine minerals and smectites (nontronite-
108 saponite; note that the identification of smectites in this study is based on X-ray diffraction
109 analyses of Hurrel, 1973). Diopsidic clinopyroxene (50-80 mol %) typically forms small
110 needles and is only rarely present as phenocrysts (Fig. 3b). Chromian spinel (up to 1.1 wt.%
111 Cr₂O₃) occurs as inclusions in olivine pseudomorphs (Fig. 3c) and in the groundmass, where it
112 commonly shows a sieve-like texture (Fig. 3d) frequently overgrown by magnetite (<0.5 apfu
113 Ti; Fig. 3c). In sample M110 and M156, some magnetites show an atoll-shaped texture with a
114 core of garnet (schorlomite-morimotoite-grossular solid solutions) or titanite (Fig. 3e).
115 Groundmass melilite is completely altered to chlorite (Fig. 3f). In some samples, purple
116 pseudomorphs after haüyne were found which consist of a very fine-grained mixture of
117 unidentified secondary minerals (Fig. 3g). Further groundmass constituents are apatite needles
118 and interstitial biotite (Fig. 3h). The latter can be distinguished into a Mg-rich ($X_{Mg}=0.7-0.8$)
119 and a Fe-rich ($X_{Mg}=0.0-0.6$) variety (Table 3). Much of the formerly glassy groundmass consists
120 of very fine-grained secondary minerals including smectites, zeolites, calcite and chlorite.

121 Based on whole-rock analyses, Wimmenauer (1951) suggested that the fine-grained
122 groundmass originally consisted of melilite, leucite and nepheline.

123 *Monchiquite* Monchiquites contain up to 3 mm large, sub- to anhedral and partly
124 rounded olivine macrocrysts (Fig. 4a) that are heavily altered and commonly replaced by
125 serpentine minerals and smectites (nontronite; Hurrell, 1973). The olivine relicts tend to higher
126 forsterite components (Fo₉₂ to Fo₈₃), higher Ni (0.28-0.37 wt.% NiO) and lower Ca (0.02-0.12
127 wt.% CaO) concentrations compared to olivine microphenocrysts of the groundmass (Fo₈₇ to
128 Fo₈₄; 0.17-0.32 wt.% NiO; 0.1-0.3 wt.% CaO; Fig.4b). Diopsidic clinopyroxene (50-80 mol %)
129 forms several mm large, euhedral to subhedral macrocrysts, which typically consist of a bright
130 sieved core, a plain transitional zone and a brown rim (Fig. 4c). Occasionally, there is an
131 irregularly shaped bright clinopyroxene generation within the sieved core (Fig. 4d) which
132 reveals the highest enstatite component (7-20 mol %) and the highest Cr concentrations (up to
133 1.65 wt.% Cr₂O₃). Green core clinopyroxenes are rare (Fig. 4e). The groundmass is dominated
134 by brown elongated clinopyroxene grains. Chromian spinel (up to 1.1 wt.% Cr₂O₃) occurs as
135 inclusions in olivine, the bright sieved clinopyroxene and within the groundmass where it
136 sometimes reveals a sieve texture. Titanomagnetite (up to 0.6 apfu Ti) only is present in the
137 groundmass. The groundmass additionally comprises partly altered glass (fine-grained zeolites,
138 calcite, chlorite, talc), apatite, small tabular kaersutite and biotite (X_{Mg}=0.4-0.6; Fig. 4b; Table
139 3). Notable are roundish areas consisting of chlorite, talc and calcite which are surrounded by
140 a zone of amphiboles and biotite grains (Fig. 4f).

141

142 *Textural and mineralogical differences of the dyke rocks depending on the host rock*
143 *(bituminous shales versus potash salt layers)*

144 Both olivine melilitites and monchiquites contain roundish textures (Fig. 5). In olivine
145 melilitites, the spheroidal to sometimes lobate textures contain diopside and perovskite at the

146 marginal zones and Hurre (1973) described additional melilite and nepheline. Hence, these
147 spheroids are interpreted as ocelli that represent immiscible droplets of a "carbonatitic" magma.
148 However, some roundish textures in these rocks do not show any indication for primary
149 magmatic mineral phases and presumably represent refilled amygdules. The spheroids in
150 monchiquites show a concentric texture and lack evidence for magmatic mineral phases.
151 Consequently, they are interpreted as amygdules (c.f. Hurre, 1973, Hurre, 1976). Other
152 roundish areas that are surrounded by seams of biotite and amphibole resemble vitreous melt
153 residues that have subsequently been altered.

154

155 *Mineralization associated with the Fish Shale* At the contact to the Fish Shale, the
156 ocelli of the olivine melilitite comprise variable amounts of irregularly distributed aggregates
157 of smectite, natrolite and small amounts of calcite (Fig. 6a&b). Biotite may additionally occur
158 at the ocelli rim. The amygdules typically consist of a marginal zone of smectite and a centre
159 with variable amounts of tabular or needle-shaped natrolite (Fig. 6b&c), euhedral analcime and
160 late-stage calcite (Fig. 6d). In some amygdules, marginal smectite is missing (Fig. 6f). Pyrite
161 may occur at the rim of the ocelli (Fig. 6e), but also replaces parts of the groundmass (Fig. 6f).

162

163 *Mineralization associated with the potash salt layer* Ocelli of the olivine
164 melilitite associated with the potash salt deposit consist of marginal smectite and sometimes
165 biotite, while the central parts of the ocelli are dominated by halite (Fig. 7a). Hematite was
166 rarely found but then typically grew on top of the smectite seam (Fig. 7b). Tabular anhydrite
167 grains and barite are particularly common in sample M67 and M156 with the latter sample
168 additionally comprising small amounts of celestite (Fig. 7b). Notably, the ocelli margins of
169 sample M156 also contain isometric garnet with inclusions of small needle-shaped
170 clinopyroxene (Fig. 7c) and dark green aegirine needles replacing diopside (Fig. 7a).

171 Furthermore, sample M156 is characterized by halite pseudomorphs after olivine (Fig. 7d).
172 Datolite grains occur in altered olivine, either associated with halite (Fig. 7d) or serpentine and
173 smectite (Fig. 7e). Amygdules generally consist of a small rim of smectite minerals, marginal
174 calcite, sometimes anhydrite, and a core of halite (Fig. 7f).

175 The amygdules of the monchiquites have a concentric succession from rim to core consisting
176 of chlorite, smectite, talc, calcite and halite. A thin seam of pyrite sometimes occurs between
177 talc and halite (Fig. 7g). In some amygdules, halite is missing (Fig. 7h), which, however, may
178 be a 3-D-effect.

179

180

DISCUSSION

181 *Crystallization conditions*

182 The anhedral olivine macrocrysts of the monchiquites (forsterite content up to 92 %)
183 show high Ni (≥ 0.3 wt % NiO) and low Ca (≤ 0.1 wt. % CaO) concentrations which are typical
184 of mantle xenocrysts (Foley *et al.*, 2013, Bussweiler *et al.*, 2015). Moreover, depth estimates
185 were obtained by the clinopyroxene barometer of Nimis & Ulmer (1998) which was applied to
186 the various clinopyroxene types of the monchiquites (Fig. 8). White core clinopyroxenes
187 demonstrate high crystallization depths roughly between 40 and 20 km (depth estimates below
188 10 km are based on altered clinopyroxenes) and may represent mantle xenocrysts or early
189 orthomagmatic crystals. Green core clinopyroxenes presumably demonstrate the involvement
190 of percolative melts at lower crustal conditions (Pilet *et al.*, 2002). The sieve-like textured
191 clinopyroxenes (Fig. 4c) and spinels indicate a partial resorption of the macrocrysts during
192 magma ascent (e.g. Qi *et al.*, 1995, Shaw *et al.*, 2006). Groundmass clinopyroxenes
193 (transitional/brown clinopyroxenes) typically show values below those expected by an
194 overburden of ~ 1000 m (approximately 0.3 kbar) which may be related to the high uncertainty

195 of the calculated values ($\sigma=1.75$ kbar) and the fact that the barometer was initially calibrated
196 for anhydrous systems. However, groundmass amphibole and mica in the monchiquites provide
197 prove of a considerable water concentration of the melt at late stage conditions. Pressure
198 estimates therefore are underestimated by ~ 1 kbar per wt.% H₂O in the melt (Nimis & Ulmer,
199 1998). Based on coexisting olivine microphenocrysts and brown clinopyroxene, the
200 groundmass temperature could be determined to approximately 950 °C (Andersen *et al.*, 1993).

201

202 *Reason for the absence of salt rock-magma interactions*

203 The large chemical contrast between the mafic dykes and the potash salts provide
204 potential for intense assimilation and metasomatic processes. However, no textural or chemical
205 differences in the magmatic mineral phases could be observed between the olivine melilitites
206 in contact to the Fish Shale and those surrounded by the potash salt layer. Note, that slightly
207 elevated Cl concentrations in biotite associated with the salt deposit (Table 3) may be caused
208 by Cl diffusion into magmatically formed biotite (high X_{Mg}) during interaction with the NaCl-
209 rich fluid, or may hint to a hydrothermal formation of biotite (low X_{Mg}). Sodium- and chlorine-
210 rich mineral phases such as sodalite, which could be expected to form during magmatic
211 interaction between a silica- undersaturated magma and salt rocks, are absent. Since spheroidal
212 textures occur in all investigated dyke rocks, independent of the surrounding host rock, we
213 exclude the possibility that the halite textures in these spheroidal structures demonstrate a salt-
214 magma mingling, or even a salt-liquid immiscibility due to intense magmatic assimilation of
215 the halite-bearing host rock. We suggest that the absence of salt rock-melt interactions is mainly
216 related to two facts:

217 i) monchiquites and olivine melilitites are characterized by the absence of OH-bearing
218 phenocryst minerals. Thus, the melts presumably had a rather low water activity during the
219 orthomagmatic stage which prevented the exsolution of large amounts of fluids and a

220 subsequent dissolution of the salt rock. Moreover, carnallite, a hydrous salt mineral that already
221 decomposes at low temperatures $<170\text{ }^{\circ}\text{C}$ (Braitsch, 1971) is relatively rare in the salt deposit
222 of Buggingen, and hence could not provide significant amounts of water (Braitsch & Herrmann,
223 1964). However, the amount of primary brines in the host rock is uncertain.

224 ii) the dyke magma cooled rapidly. Knipping (1989) calculated the temperature for salt rocks
225 that were intruded by basaltic magmas in the Werra Fulda region. He assumed a dyke
226 thicknesses of 0.3, 1.0 and 1.8 m, a magma temperature of $1150\text{ }^{\circ}\text{C}$, and a wall rock temperature
227 of $50\text{ }^{\circ}\text{C}$. These parameters can also be applied to our study, where the dike widths vary between
228 a few cm to a maximum of 4 m (Esslinger, 1976), the magma temperature is estimated to be
229 above $950\text{ }^{\circ}\text{C}$, and the reported air temperature of the mine at 1000 m depth was about $50\text{ }^{\circ}\text{C}$
230 (Markl, 2017; we assume little erosion of the overburden since the intrusion of the dyke rocks).
231 Based on these input parameters, the calculated wall-rock temperature at the contact to the dyke
232 always remained below $790\text{ }^{\circ}\text{C}$, which prevented melting of halite and even sylvite (melting
233 point of ~ 800 and $\sim 790\text{ }^{\circ}\text{C}$ respectively). This coincides with the straight contact between the
234 salt rock and the dyke at Buggingen which does not indicate partial melting of the wall-rock.
235 However, at branching dykes, between (almost) contemporaneously emplaced intrusions, or if
236 salt fragments got entrapped into the magma, heat could act from more than one side and the
237 salt may locally have reached the melting temperature, as suggested by Koch & Vogel (1980)
238 in the Werra-Fulda region in Central Germany. From the same region, Schofield *et al.* (2014)
239 described some peperite structures at the contact between salt and dyke rocks which also
240 indicate a magmatic interaction. However, this process was caused by heating and dehydration
241 of carnallite, which is relatively rare at Buggingen (Braitsch & Herrmann, 1964).

242

243 *Fluid conditions*

244 The minimum temperature of the fluid responsible for the precipitation of halite in
245 amygdules and ocelli was estimated to ~50 °C based on the current temperature of the mine at
246 ~1000 m depth. Evidence for a somewhat higher minimum temperature of a hydrothermal event
247 is shown by sample M156, where the presence of garnet and aegirine-rich clinopyroxene in the
248 ocelli illustrates temperatures of at least 200 °C (Easton *et al.*, 1977, Decarreau *et al.*, 2004).
249 Moreover, the core of atoll-shaped magnetite in sample M110 and M156 consists of garnet or
250 titanite, and may demonstrate a replacement of Ti-rich magnetite by relatively hot fluids that
251 penetrated the grain along cracks. However, a direct link between these relatively hot fluids and
252 the halite-bearing spheroids is uncertain. While textural observations of Wimmenauer (1951)
253 suggest that the salt layer in contact to the dyke rock is almost completely impoverished in KCl,
254 Braitsch *et al.* (1964) demonstrated by chemical analyses that the contact zones still contain
255 high KCl concentrations of up to 40 wt.%. We therefore suggest that the hydrothermal fluid
256 contained a mixture of dissolved KCl and NaCl. Due to the high variations of the KCl
257 concentration in the salt layers (between 3 and 40 wt.% KCl; Braitsch *et al.*, 1964), and the high
258 temperature dependence of the KCl solubility (Braitsch, 1971), we omit a quantitative
259 calculation for the salt content of the fluid.

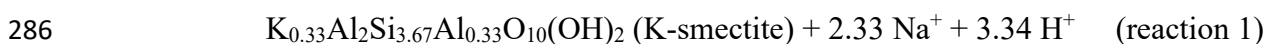
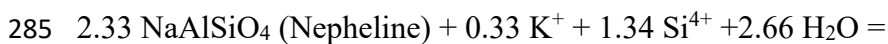
260 The hydrothermal fluid that interacted with the olivine melilitites associated with the Fish Shale
261 most likely equilibrated in this bituminous, pyrite-bearing marl (Böcker & Littke, 2014,
262 Eichertopf *et al.*, 2017). This is consistent with the high amounts of zeolite, analcime and
263 calcite in the ocelli and amygdule fillings, as well as with the presence of pyrite in the
264 groundmass.

265

266 *The crystallization process for halite-bearing amygdules and ocelli*

267 Evaporation can be excluded as the precipitation mechanism for the hydrothermal
268 formation of halite because the mafic dyke rocks were sampled at depth. In the following, we
269 discuss a possible crystallization process for the halite-bearing spheroids (Fig. 9a). Due to the
270 complex cation incorporation and adsorption capacity of smectites (e.g. Klopogge *et al.*, 1999),
271 the importance of kinetics (Gysi & Stefánsson, 2011) as well as the large number of uncertain
272 parameters (e.g. fluid composition; fluid/dyke rock temperature; groundmass mineralogy of
273 pristine dyke rocks), we chose not to construct a thermodynamic model of the fluid evolution.
274 We present a geologically sound, but qualitative approach instead (Fig. 9b).

275 Assuming that the fine-grained groundmass of the monchiquites and the olivine melilitites
276 mainly consisted of nepheline (as being typical of most silica-undersaturated primitive rocks),
277 the hydrothermal alteration to clay minerals \pm zeolites (Fig. 9a-1) should result in an increase
278 of the sodium concentration of the fluid. Indeed, whole-rock analyses (after subtracting halite
279 and anhydrite) of the heavily altered dyke rocks at Buggingen have lower Na concentrations
280 than relatively fresh rock samples (Hurrel, 1973). In contrast, the K concentration of
281 hydrothermally strongly overprinted rock samples is higher compared to the rather fresh dyke
282 rocks. This also conforms with chemical analyses of the clay minerals, which typically consist
283 of K-bearing smectites (Hurrel, 1973). Such a groundmass alteration process typically releases
284 more Na than it consumes K (reaction 1),



287 The uptake of sodium into a salt-saturated fluid should immediately lead to the precipitation of
288 halite. However, as no halite was observed in the groundmass (note that the thin sections were
289 prepared completely dry, hence excluding a preparation-based salt dissolution), we suggest that

290 the initial fluid was not saturated with halite before penetrating the dyke rock, possibly due to
291 mixing and diluting with meteoric water. During interaction with the groundmass, the Na
292 concentration of the fluid steadily increased (while the K concentration decreased) but
293 obviously did not yet reach NaCl saturation. As soon as the fluid hit an amygdule, smectites
294 formed at the rim and the fluid became trapped (Fig. 9a-2). The Na concentration of the fluid
295 further increased during interaction with the marginal zone while the salt-rich fluids finally
296 desiccated in the vugs due to swelling of smectites (e.g. Usuki *et al.*, 1993). Consequently, NaCl
297 saturation was reached and halite precipitated in the vugs (Fig. 9a-3). In the amygdules, the
298 crystallization of talc additionally removed H₂O from the fluid and drove it towards NaCl
299 saturation. A similar halite precipitation process can be assumed for the ocelli. In two olivine
300 melilitites, Hurre (1973) found ocelli with only partly altered nepheline and a core of calcite,
301 which possibly demonstrates the primary magmatic ocelli filling. We assume that calcite
302 dissolved during interaction with the fluid, while smectites formed at the ocelli rim by alteration
303 of the marginal mineral assemblage of the ocelli (Fig. 10). Subsequent desiccation of the
304 trapped fluids by clay mineral swelling led to the crystallization of halite.

305 Smectite and zeolite formation in the groundmass presumably cemented possible fluid paths
306 (Fig. 9a-3), which would indicate that the salt-rich spheroids formed by a single hydrothermal
307 event. In amygdules, halite comprises up to ~12-20 vol.%, which is in the same order of
308 magnitude as the 16.5 vol.% halite that can precipitate from an NaCl-saturated (~30 wt.% NaCl)
309 fluid at ~50 °C, assuming equimolar Na⁺ and Cl⁻ (Driesner & Heinrich, 2007). As a
310 consequence, we suggest that the fluid was near NaCl saturation when it got trapped in the vugs
311 and finally desiccated by clay mineral swelling. Due to the uncertainty about the dissolved
312 amount of former mineral phases in the ocelli, we omit similar calculations for this spheroid
313 type.

314 The effect of clay mineral swelling is also illustrated by the mineral proportions in the
315 ocelli/amygdules. Based on the simplified assumption, that the OH-bearing mineral phases of
316 the ocelli/amygdules have an average volume of 140 cm³/mol (halite 30 cm³/mol; Blanc *et al.*,
317 2012 and references therein), and only contain 1 OH-group per formula unit (no interlayer
318 water), the observed amount of smectite in the amygdules/ocelli (~85 vol.%) that formed due
319 to the interaction between the spheroid rim and the NaCl-saturated fluid, is approximately 9
320 times too low. The absence of a prominent clay rim in some amygdules (Fig. 7f) implies that
321 nearby clay minerals of the groundmass additionally bound H₂O in their interlayers. Indeed, the
322 lack of OH-bearing minerals at the amygdule rims is restricted to the olivine melilitites, which
323 in contrast to the monchiquites, contain much more smectite minerals in the groundmass.

324

325 *Crystallization of datolite*

326 Datolite [CaB(SiO₄)(OH)] only occurs in one olivine melilitite sample (M156) as
327 pseudomorphs after olivine, either associated with serpentine and smectite (Fig. 7e) or halite
328 (Fig. 7d), but it is never present within ocelli. The required boron presumably derived from
329 ocean-related sedimentary rocks (e.g. Leeman & Sisson, 1996, Garrett, 1998). The salt rocks at
330 Buggingen are unlikely to represent the source of the boron since the concentrations are very
331 low (< 10 ppm; Braitsch *et al.*, 1964), but the clay-rich layers of the salt horizon contain up to
332 600 ppm boron. Braitsch *et al.* (1964) did not believe in this boron source in the country rocks
333 and suggested a magmatic origin since they could not detect significant variations in the boron
334 content between the clay-rich layers in contact to the dyke rocks and those further away.
335 However, we suggest that boron-leaching of the wall rock would have too little effect to be
336 detectable, since boron is adsorbed by clays and should have been balanced again by circulating
337 fluids over the course of the years. The high Ca concentration of the fluid required for the

338 crystallization of datolite most likely derived from the anhydrite-rich parts of the host rock,
339 which is consistent with high amounts of anhydrite in ocelli of sample M156 (Fig. 7b).

340 The formation of datolite presumably was caused by a two-step process. This is in accordance
341 with the restriction of datolite to only one rock sample of the present study (it has been found
342 more often in other sample suites, though; Braitsch *et al.*, 1964, Markl, 2017), since clay
343 mineral and zeolite precipitation generally cement the fluid paths. However, in case of sample
344 M156, new cracks possibly formed due to tectonic stress (Wimmenauer, 1951). The first fluid
345 batch equilibrated with the dyke rock by the alteration of olivine to serpentine and Mg-rich
346 smectites (e.g. saponite; Hurrell, 1973). Boron thereby was incorporated into serpentine (e.g.
347 Pabst *et al.*, 2011) and was adsorptively bound in the clay minerals (e.g. Gu & Lowe, 1990,
348 Goldberg *et al.*, 1993, Karahan *et al.*, 2006). A second fluid batch which then was in
349 disequilibrium with the olivine alteration phases dissolved serpentine and Mg-rich smectite.
350 The high amounts of released boron and silica finally led to the crystallization of datolite grains.
351 In contrast to the ocelli and amygdules, anhydrite is absent as presumably all available Ca was
352 incorporated in datolite.

353

354

SUMMARY AND CONCLUSION

355 No evidence for a magmatic interaction between the primitive dyke magma and the salt
356 layers at Buggingen in SW Germany was observed. This is presumably related to (i) relatively
357 dry orthomagmatic melt conditions and to (ii) a too fast cooling of the dyke. Instead, halite-
358 bearing spheroids in the olivine melilitites and the monchiquites represent amygdules,
359 respectively hydrothermally altered ocelli. Low temperature NaCl-bearing fluids from the
360 surrounding salt layers reacted with the dyke rock groundmass, thereby forming smectites and
361 zeolites. During alteration of the groundmass, the NaCl concentration of the fluid steadily
362 increased but remained below NaCl saturation. As soon as the fluid hit a vug, smectites formed

363 at the margin and the fluid got trapped. Clay mineral swelling gradually consumed water from
364 the fluid and finally led to the precipitation of halite. The presence of datolite is restricted to
365 pseudomorphs after olivine. Serpentine/smectite breakdown provided the required Si and boron
366 for its formation.

367

368

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374

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TABLE 1. REPRESENTATIVE EPMA ANALYSES OF OLIVINE AND CLINOPYROXENE

Rock type Sample # Association	Monchiquite						Olivine mellilitite					
	Olivine			Clinopyroxene			Potash salt			Clinopyroxene		
	Large-anhedral core	Small-euhedral rim	White core	Sieved	Transitional	Brown	Green core	Groundmass	Ocellus	Groundmass	Ocellus	Ocellus
Mineral	M164	M63	M13	8111	M164	M13	M63	M13	M34	M156	M109	M110
Location	core	rim	core	rim	White core	Sieved	Transitional	Brown	Groundmass	Ocellus	Groundmass	Ocellus
wt. %	41.74	41.16	40.87	40.83	52.76	49.94	42.33	47.39	46.45	46.72	47.62	47.71
SiO ₂	na	na	na	na	0.61	1.42	4.69	3.03	2.87	3.46	2.45	3.36
TiO ₂	na	na	na	na	bdl	bdl	0.04	0.03	0.06	0.06	bdl	0.14
ZrO ₂	bdl	0.03	0.07	0.05	3.88	4.70	9.67	6.05	5.94	5.02	5.10	4.71
Al ₂ O ₃	na	na	na	na	0.80	1.12	0.23	bdl	bdl	bdl	0.10	bdl
Cr ₂ O ₃	7.50	10.74	11.83	11.67	1.23	2.43	5.67	2.51	4.83	4.28	4.88	3.52
Fe ₂ O ₃ *	0.14	0.18	0.16	0.16	2.93	2.33	2.41	4.71	2.54	3.45	1.63	2.57
MnO	50.52	47.67	47.25	46.65	0.14	0.06	0.06	0.11	0.09	0.15	0.12	0.16
MgO	0.34	0.31	0.20	0.27	17.26	14.88	11.50	13.02	13.20	12.90	13.83	13.86
NiO	0.02	0.12	0.18	0.19	na	na	na	na	na	na	na	na
CaO	na	na	na	na	20.11	23.04	23.10	22.62	23.69	23.41	24.32	23.78
Na ₂ O	na	na	na	na	0.87	0.55	0.49	0.52	0.36	0.51	0.34	0.48
K ₂ O	na	na	na	na	bdl	bdl	bdl	bdl	bdl	0.02	bdl	bdl
Total	100.27	100.20	100.55	99.82	100.59	100.48	100.19	99.99	100.03	99.98	100.40	100.30
<i>Formula based on 4 oxygens (olivine), respectively 4 cations and 6 oxygens (clinopyroxene)</i>												
Si	1.01	1.01	1.00	1.01	1.90	1.83	1.59	1.77	1.73	1.75	1.76	1.77
Ti	na	na	na	na	0.02	0.04	0.13	0.08	0.08	0.10	0.07	0.09
Zr	na	na	na	na	bdl	bdl	0.00	0.00	0.00	0.00	bdl	0.00
Al	0.00	0.00	0.00	0.00	0.16	0.20	0.43	0.27	0.26	0.22	0.22	0.21
Cr	na	na	na	na	0.02	0.03	0.01	bdl	bdl	bdl	0.00	bdl
Fe ³⁺	na	na	na	na	0.03	0.07	0.16	0.07	0.14	0.12	0.14	0.10
Fe ²⁺	0.15	0.22	0.24	0.24	0.09	0.07	0.08	0.15	0.08	0.11	0.05	0.08
Mn	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01
Mg	1.82	1.74	1.73	1.72	0.93	0.81	0.64	0.72	0.73	0.72	0.76	0.77
Ni	0.01	0.01	0.00	0.01	na	na	na	na	na	na	na	na
Ca	0.00	0.00	0.00	0.00	0.78	0.90	0.93	0.90	0.95	0.94	0.96	0.94
Na	na	na	na	na	0.06	0.04	0.04	0.04	0.03	0.04	0.02	0.03
K	na	na	na	na	bdl	bdl	bdl	bdl	bdl	0.00	bdl	bdl
Total cations	2.00	2.00	2.00	2.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00

na: not analyzed; bdl: below detection limit; *Fe₂O₃-FeO ratio for clinopyroxene was calculated.

TABLE 2. REPRESENTATIVE EPMA ANALYSES OF SPINEL GROUP MINERALS

Rock type Sample # Association	Monchiquite				Olivine melilitite														
	M164		M13		M34		M156		M34		M67		M89		M110		M110		
	Chromite		Spinel		Potash salt		Chromite		Spinel		Potash salt		Magnetite (altered)		Spinel		Magnetite (altered)		
Mineral	Chromite	Spinel	Magnetite	Chromite	Spinel	Spinel	Spinel	Chromite	Spinel	Spinel	Magnetite	Magnetite (altered)	Magnetite (altered)	Spinel	Magnetite	Magnetite (altered)	Magnetite (altered)	Magnetite (altered)	
wt.%	0.10	0.11	0.13	0.08	0.11	0.13	0.13	0.08	0.11	0.13	0.13	0.09	0.09	0.11	0.19	0.15	0.15	0.15	
SiO ₂	4.05	1.70	17.86	1.69	4.75	12.10	12.10	1.69	4.75	12.10	12.10	8.53	8.53	4.67	7.28	12.10	12.10	12.10	
TiO ₂	9.52	26.13	6.27	11.76	16.42	6.68	6.68	11.76	16.42	6.68	6.68	0.34	0.34	18.03	9.58	0.64	0.64	0.64	
Al ₂ O ₃	0.19	0.15	0.38	0.08	0.12	0.12	0.12	0.08	0.12	0.12	0.12	0.30	0.30	0.11	0.19	0.25	0.25	0.25	
V ₂ O ₃	40.25	33.04	0.00	43.92	19.37	0.07	0.07	43.92	19.37	0.07	0.07	bdl	bdl	20.51	16.58	bdl	bdl	bdl	
Cr ₂ O ₃	15.13	27.34	29.20	13.86	26.74	41.88	41.88	13.86	26.74	41.88	41.88	51.39	51.39	24.95	30.46	43.61	43.61	43.61	
Fe ₂ O ₃ *	20.95	24.61	40.53	16.97	20.83	29.50	29.50	16.97	20.83	29.50	29.50	38.14	38.14	19.88	27.92	38.90	38.90	38.90	
FeO*	0.17	0.10	0.65	0.13	0.28	0.52	0.52	0.13	0.28	0.52	0.52	0.59	0.59	0.24	0.88	2.73	2.73	2.73	
MnO	10.46	13.99	4.93	11.93	11.40	8.72	8.72	11.93	11.40	8.72	8.72	0.11	0.11	12.36	7.27	0.09	0.09	0.09	
MgO	0.28	0.22	0.03	0.10	0.10	0.07	0.07	0.10	0.10	0.07	0.07	0.02	0.02	0.11	0.03	0.02	0.02	0.02	
NiO	0.09	0.11	0.08	0.09	0.07	0.06	0.06	0.09	0.07	0.06	0.06	0.05	0.05	0.07	0.12	0.46	0.46	0.46	
ZnO	101.18	101.09	100.05	100.59	100.20	99.85	99.85	100.59	100.20	99.85	99.85	99.56	99.56	101.03	100.48	98.94	98.94	98.94	
Total																			
<i>Formula based on 3 cations and 4 oxygens</i>																			
Si	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.01	0.01	0.01	
Ti	0.10	0.04	0.47	0.04	0.11	0.31	0.31	0.04	0.11	0.31	0.31	0.25	0.25	0.11	0.19	0.35	0.35	0.35	
Al	0.37	0.92	0.26	0.45	0.62	0.27	0.27	0.45	0.62	0.27	0.27	0.02	0.02	0.67	0.38	0.03	0.03	0.03	
V	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	
Cr	1.05	0.78	0.00	1.12	0.49	0.00	0.00	1.12	0.49	0.00	0.00	bdl	bdl	0.51	0.45	bdl	bdl	bdl	
Fe ³⁺	0.37	0.21	0.78	0.34	0.65	1.09	1.09	0.34	0.65	1.09	1.09	1.48	1.48	0.59	0.78	1.26	1.26	1.26	
Fe ²⁺	0.58	0.41	1.20	0.46	0.56	0.85	0.85	0.46	0.56	0.85	0.85	1.22	1.22	0.52	0.80	1.25	1.25	1.25	
Mn	0.00	0.00	0.02	0.00	0.01	0.02	0.02	0.00	0.01	0.02	0.02	0.02	0.02	0.01	0.03	0.09	0.09	0.09	
Mg	0.51	0.62	0.26	0.58	0.55	0.45	0.45	0.58	0.55	0.45	0.45	0.01	0.01	0.58	0.37	0.01	0.01	0.01	
Ni	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	
Zn	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.01	0.01	
Total cations	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	3.00	

bdl: below detection limit; *Fe₂O₃-FeO ratio was calculated.

TABLE 3. REPRESENTATIVE EPMA ANALYSES OF AMPHIBOLE AND MICA

Rock type	Monchiquite						Olivine melilitite					
	Amphibole			Mica			Potash salt			Mica		
	M63	M63	8111	M13	M63	M34	M156	M156	M156	M89	M110	M89
Sample #	M63	M63	8111	M13	M63	M34	M156	M156	M156	M89	M110	M89
Association	Amphibole			Mica			Potash salt			Mica		
Mineral	Amphibole			Mica			Potash salt			Mica		
Location	Amphibole			Mica			Potash salt			Mica		
wt.%	Groundmass			Groundmass			Ocellus			Groundmass		
SiO ₂	36.53	36.07	36.20	33.28	33.50	35.59	35.77	36.77	36.77	33.36	36.43	37.99
TiO ₂	6.85	7.41	7.87	8.35	6.82	5.50	5.54	2.97	2.97	7.19	5.84	6.60
ZrO ₂	0.05	0.04	0.04	bdl	0.02	0.02	0.02	0.02	0.02	bdl	bdl	bdl
Al ₂ O ₃	13.75	15.00	14.02	16.23	15.58	14.11	15.05	6.97	6.97	14.70	14.26	7.44
Cr ₂ O ₃	bdl	0.03	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Fe ₂ O ₃ *	7.12	8.99	7.96	13.87	19.16	12.15	10.55	31.05	31.05	12.26	9.29	23.41
FeO*	9.74	4.16	5.94	0.14	0.19	0.25	0.18	0.63	0.63	0.25	0.14	0.65
MnO	0.30	0.16	0.26	0.14	0.19	0.25	0.18	0.63	0.63	0.25	0.14	0.65
MgO	8.56	10.92	10.38	12.66	9.79	16.75	17.48	8.39	8.39	15.07	18.92	9.24
CaO	11.38	11.45	11.44	0.11	0.09	0.04	0.07	0.22	0.22	0.24	0.04	0.10
BaO	0.33	0.45	0.51	3.67	3.21	2.27	3.26	1.00	1.00	4.81	3.32	0.59
Na ₂ O	2.54	2.42	2.21	0.94	0.80	0.48	0.48	0.15	0.15	0.43	0.55	0.33
K ₂ O	1.61	1.73	1.87	7.24	7.66	8.68	8.39	8.87	8.87	7.87	8.41	9.09
F	0.17	0.22	0.13	0.35	0.34	0.45	0.64	0.29	0.29	0.72	0.98	0.64
Cl	0.10	0.09	0.08	0.04	0.11	0.04	0.06	0.04	0.04	bdl	0.02	bdl
H ₂ O**	0.73	0.52	0.53	3.65	3.58	3.70	3.65	3.54	3.54	3.49	3.54	3.39
-O≡F, Cl	0.09	0.11	0.07	0.15	0.17	0.20	0.28	3.84	3.84	0.30	0.41	0.27
Total	99.66	99.54	99.36	100.38	100.67	99.84	100.86	100.79	100.79	100.09	101.32	99.18
<i>Formula based on 16 cations and 23 oxygens (Amphibole), respectively 22 oxygens (Mica)</i>												
Si	5.64	5.49	5.56	5.03	5.17	5.34	5.29	5.96	5.96	5.10	5.33	6.00
Ti	0.80	0.85	0.91	0.95	0.79	0.62	0.62	0.36	0.36	0.83	0.64	0.78
Zr	0.00	0.00	0.00	bdl	0.00	0.00	0.00	0.00	0.00	bdl	bdl	bdl
Al	2.50	2.69	2.54	2.89	2.83	2.49	2.63	1.33	1.33	2.65	2.46	1.38
Cr	bdl	0.00	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Fe ³⁺	0.82	1.01	0.90	0.00	0.00	1.52	1.31	4.21	4.21	1.57	1.14	3.09
Fe ²⁺	1.24	0.52	0.75	1.75	2.47	0.03	0.02	0.09	0.09	0.03	0.02	0.09
Mn	0.04	0.02	0.03	0.02	0.02	0.03	0.02	0.02	0.02	0.03	0.02	0.09
Mg	1.97	2.48	2.38	2.85	2.25	3.74	3.86	2.03	2.03	3.43	4.13	2.17
Ca	1.88	1.87	1.88	0.02	0.02	0.01	0.01	0.04	0.04	0.04	0.00	0.02
Ba	0.02	0.03	0.03	0.22	0.19	0.13	0.19	0.06	0.06	0.29	0.19	0.04
Na	0.76	0.71	0.66	0.28	0.24	0.14	0.14	0.05	0.05	0.13	0.16	0.10
K	0.32	0.34	0.37	1.40	1.51	1.66	1.58	1.83	1.83	1.53	1.57	1.83
Total cations	16.00	16.00	16.00	15.41	15.50	15.69	15.64	15.95	15.95	15.59	15.65	15.50
F	0.08	0.11	0.06	0.16	0.16	0.21	0.30	0.15	0.15	0.35	0.46	0.32
Cl	0.03	0.02	0.02	0.01	0.03	0.01	0.01	0.01	0.01	bdl	0.00	0.00
OH	0.75	0.52	0.54	3.82	3.81	3.78	3.69	3.84	3.84	3.65	3.54	3.68
O	1.15	1.35	1.38	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00
Total anions	2.00	2.00	2.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00	4.00

bdl: below detection limit; *Fe2O3-FeO ratio was calculated; **H2O was calculated.

TABLE 4. REPRESENTATIVE EPMA ANALYSES OF GARNET, TITANITE AND PEROVSKITE

Rock type Sample # Association Mineral Location	Olivine melilitite								
	M156			M110			M156		
	Potash salt		Fish Shale	Potash salt		Fish Shale	Potash salt		Fish Shale
	Garnet			Perovskite			Titanite		
	Ocellus	Groundmass	Ocellus	Groundmass	Ocellus	Groundmass			
wt. %									
SiO ₂	35.77	33.99	31.74	bdl	0.07	bdl	31.90	30.52	32.66
TiO ₂	3.37	5.89	10.81	54.31	55.53	54.31	30.71	34.10	28.61
ZrO ₂	0.17	bdl	bdl	bdl	bdl	bdl	0.24	0.23	bdl
Al ₂ O ₃	8.52	2.70	4.59	bdl	0.07	0.05	2.00	0.21	5.48
Fe ₂ O ₃ *	15.68	20.83	13.10	1.18	0.82	1.07	4.28	3.65	1.78
FeO*	0.35	1.65	4.34						
MnO	0.17	0.19	0.06	bdl	0.04	0.03	0.05	0.03	0.02
MgO	0.90	0.78	0.25	bdl	0.07	0.04	0.36	0.05	0.28
CaO	34.28	33.03	32.79	36.82	38.81	36.81	26.53	26.82	27.55
Na ₂ O	0.03	0.12	0.19	1.38	0.85	1.12	0.08	0.35	0.33
Nb ₂ O ₅	0.13	0.15	bdl	2.13	1.65	2.01	bdl	1.72	bdl
Ta ₂ O ₅	bdl	bdl	bdl	0.08	bdl	0.07	bdl	0.07	bdl
SrO	bdl	bdl	bdl	0.81	0.97	1.13	bdl	0.55	bdl
La ₂ O ₃	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Ce ₂ O ₃	bdl	0.08	bdl	0.37	bdl	0.64	bdl	bdl	bdl
Pr ₂ O ₃	bdl	bdl	bdl	0.19	0.07	0.19	bdl	bdl	bdl
Nd ₂ O ₃	bdl	bdl	bdl	0.17	0.06	0.31	bdl	bdl	bdl
Sm ₂ O ₃	bdl	bdl	bdl	0.09	bdl	0.07	0.04	bdl	bdl
Y ₂ O ₃	bdl	bdl	0.09	bdl	bdl	bdl	bdl	bdl	bdl
ThO ₂	bdl	bdl	bdl	0.11	0.06	0.20	bdl	bdl	bdl
F	0.04	bdl	bdl	0.34	0.15	0.12	bdl	bdl	0.03
Cl	bdl	bdl	0.02	bdl	bdl	bdl	0.36	bdl	0.03
H ₂ O**							3.23	1.66	3.65
-O≡F,Cl	0.02	bdl	0.00	0.14	0.06	0.05	0.00	0.00	0.02
Total	99.39	99.39	97.97	97.82	99.15	98.11	99.78	99.94	100.41
<i>Formula based on 8 cations and 12 oxygens (Garnet), 3 oxygens (Perovskite) or 1 Si cation (Titanite)</i>									
Si	2.89	2.84	2.68	bdl	0.00	bdl	1.00	1.00	1.00
Ti	0.21	0.37	0.69	0.97	0.97	0.97	0.72	0.84	0.66
Zr	0.01	bdl	bdl	bdl	bdl	bdl	0.00	0.00	bdl
Al	0.81	0.27	0.46	bdl	0.00	0.00	0.07	0.01	0.20
Fe ³⁺	0.95	1.31	0.83	0.01	0.01	0.01	0.10	0.09	0.04
Fe ²⁺	0.02	0.12	0.31						
Mn	0.01	0.01	0.00	bdl	0.00	0.00	0.00	0.00	0.00
Mg	0.11	0.10	0.03	bdl	0.00	0.00	0.02	0.00	0.01
Ca	2.97	2.96	2.97	0.93	0.96	0.93	0.89	0.94	0.90
Na	0.01	0.02	0.03	0.06	0.04	0.05	0.00	0.02	0.02
Nb	0.00	0.01	bdl	0.02	0.02	0.02	0.00	0.03	bdl
Ta	bdl	bdl	bdl	0.00	bdl	0.00	0.00	0.00	bdl
Sr	bdl	bdl	bdl	0.01	0.01	0.02	0.00	0.01	bdl
La	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Ce	bdl	0.00	bdl	0.00	bdl	0.01	bdl	bdl	bdl
Pr	bdl	bdl	bdl	0.00	0.00	0.00	bdl	bdl	bdl
Nd	bdl	bdl	bdl	0.00	0.00	0.00	bdl	bdl	bdl
Sm	bdl	bdl	bdl	0.00	bdl	0.00	0.00	bdl	bdl
Y	bdl	bdl	0.00	bdl	bdl	bdl	bdl	bdl	bdl
Th	bdl	bdl	bdl	0.00	0.00	0.00	bdl	bdl	bdl
Total cations	8.00	8.00	8.00	2.02	2.01	2.01	2.82	2.95	2.83
F	0.01	bdl	bdl	0.01	0.01	0.00	bdl	bdl	0.00
Cl	bdl	bdl	0.00				0.02	bdl	0.00
OH							0.72	0.37	0.79
O	11.99	12.00	12.00	2.99	2.99	3.00	4.26	4.63	4.21
Total anions	12.00	12.00	12.00	3.00	3.00	3.00	5.00	5.00	5.00

bdl: below detection limit; *Fe₂O₃-FeO ratio was calculated; **H₂O was calculated.

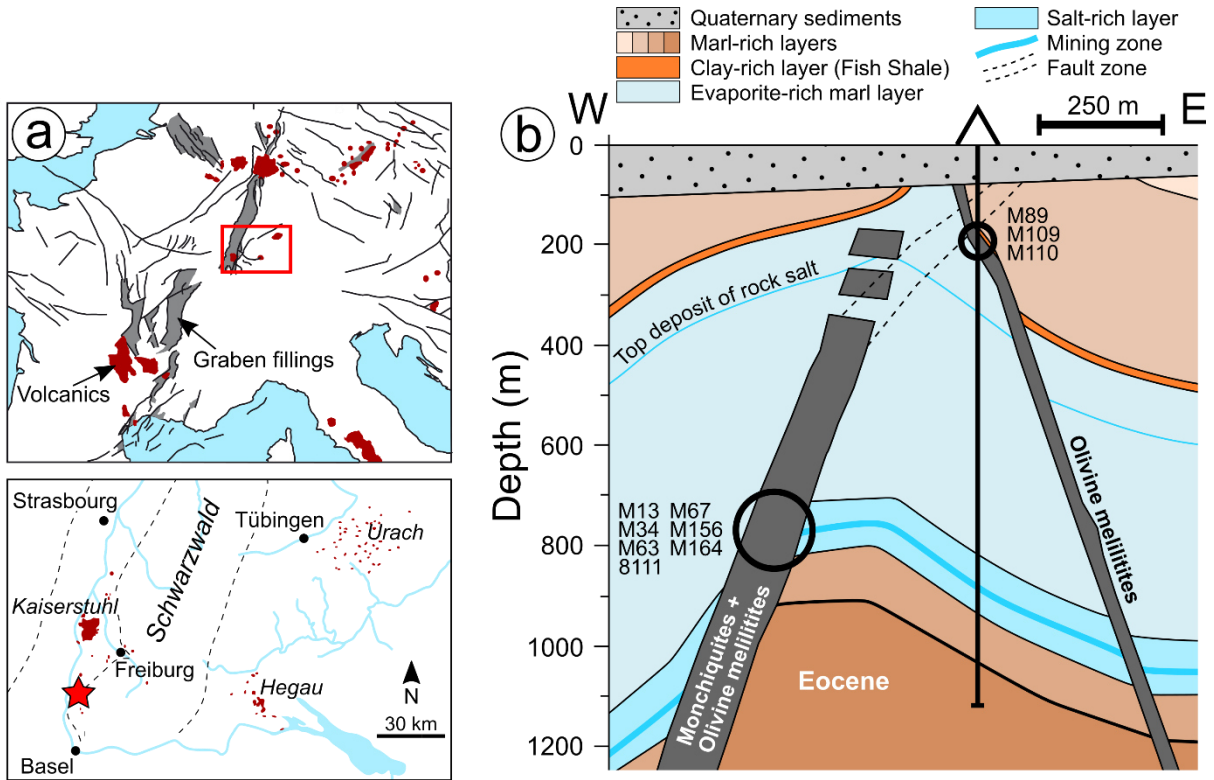


Fig. 1: Geology of Buggingen. A) Overview map of the European Cenozoic Volcanic Province and the Cenozoic magmatic occurrences in southwest Germany (modified after Dèzes *et al.*, 2004). Buggingen is located at the red star. B) Cross-section showing the geology of Buggingen and the sample sites (modified after Esslinger, 1976).

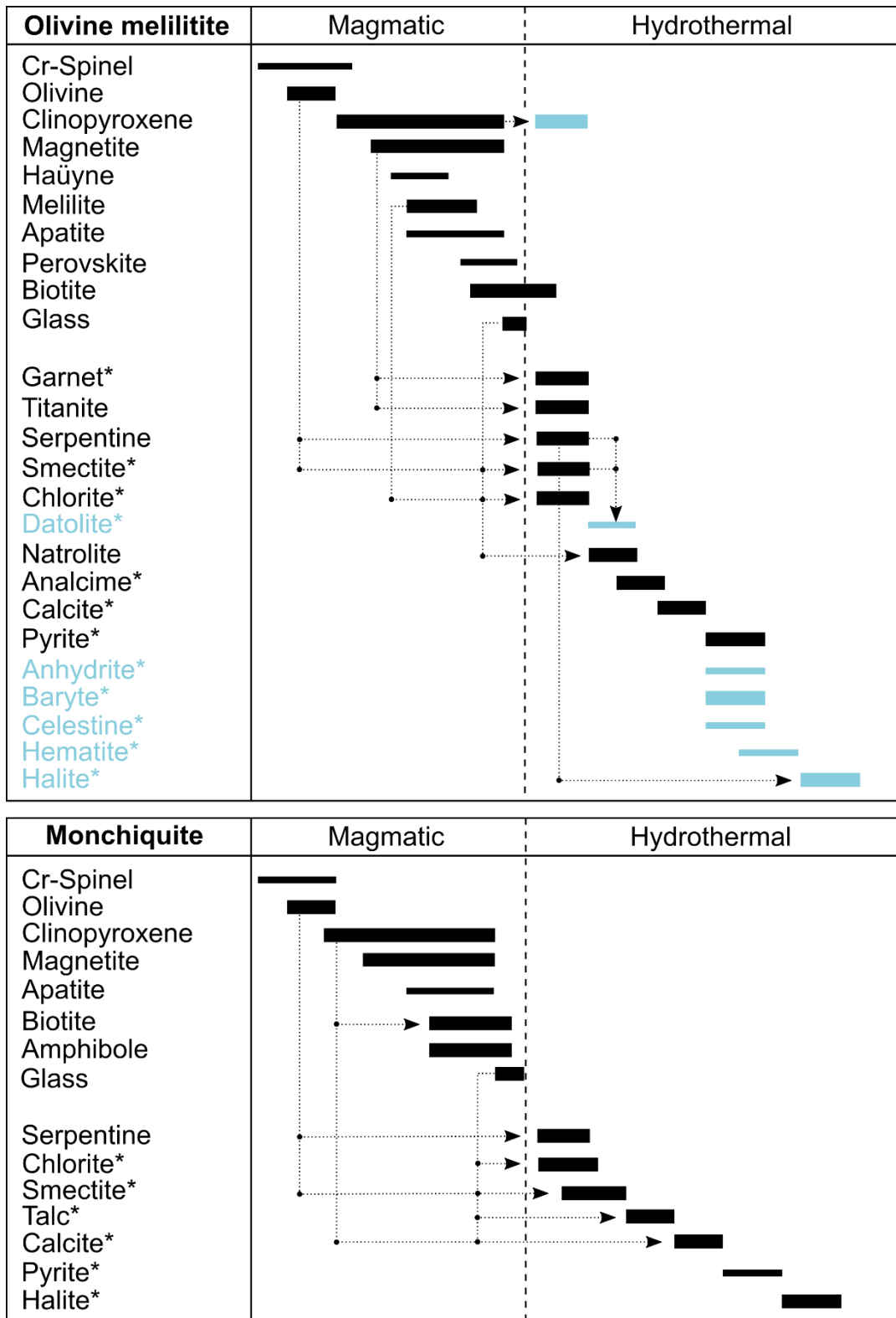


Fig. 2: Paragenetic scheme for the olivine melilitite and the monchiquites from Buggingen. Alteration processes are indicated by arrows. Mineral phases of the olivine melilitite that only occur in association with the salt layer are indicated in blue. Mineral phases highlighted by the star are present in ocelli/amygdules.

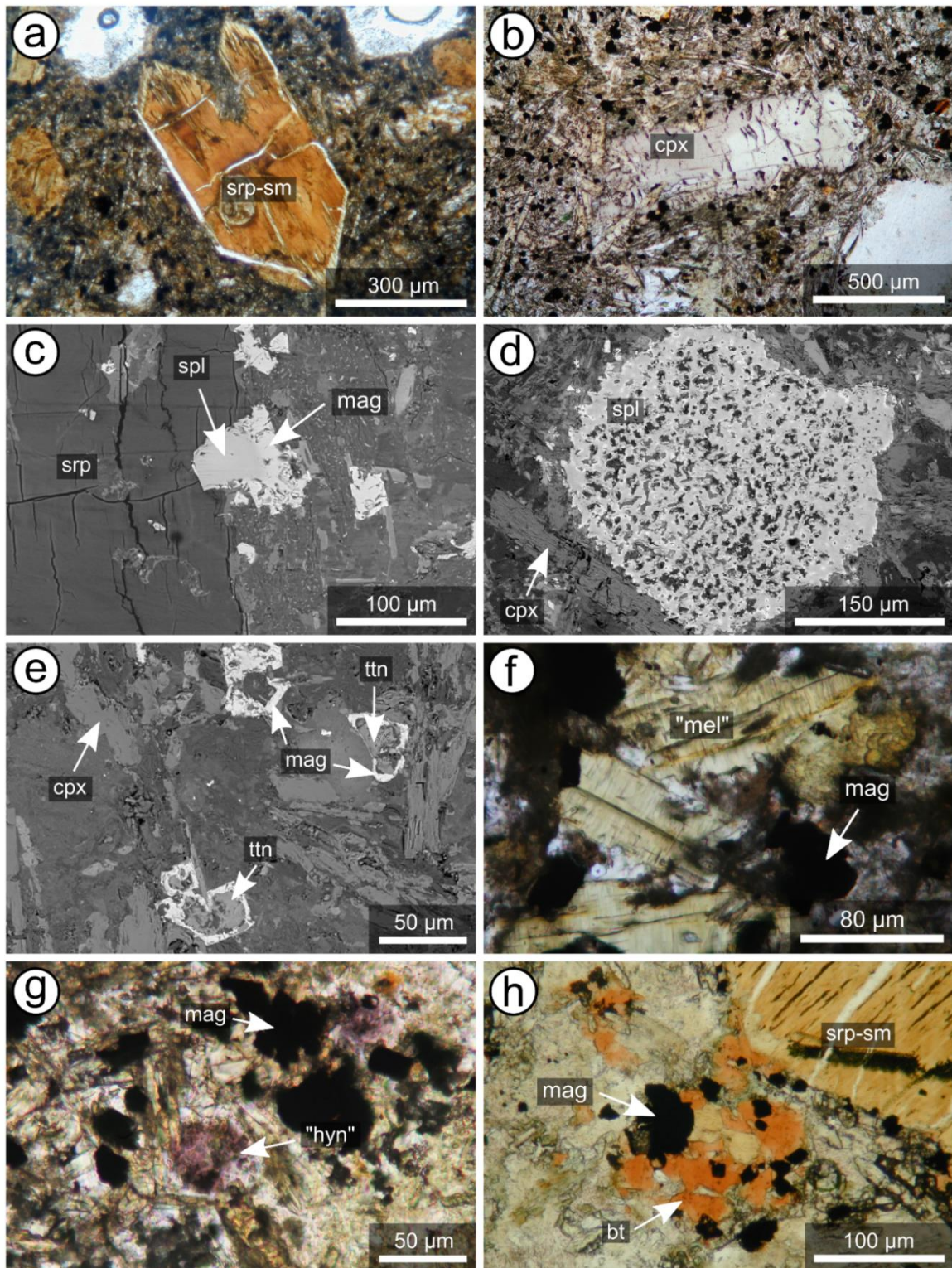


Fig. 3: Mineral textures in the olivine melilitites from Buggingen (transmitted light microscopy and BSE images [c,d,e]). a) Euhedral olivine macrocryst totally replaced by serpentine (srp) and smectite (sm) minerals. b) Clinopyroxene (cpx) macrocryst in a groundmass consisting of plenty clinopyroxene needles. c) Spinel (spl) at the rim of a serpentinized olivine. Those parts of the spinel that are not enclosed by former olivine are surrounded by magnetite (mag). d) Spinel showing a sieve texture. e) Atoll-shaped magnetite with a core of titanite (ttn). f) Melilite (mel) being totally altered to chlorite. g) Purple pseudomorphs after haüyne (hyn) consisting of a mixture of fine-grained secondary minerals. e) Interstitial biotite (bt).

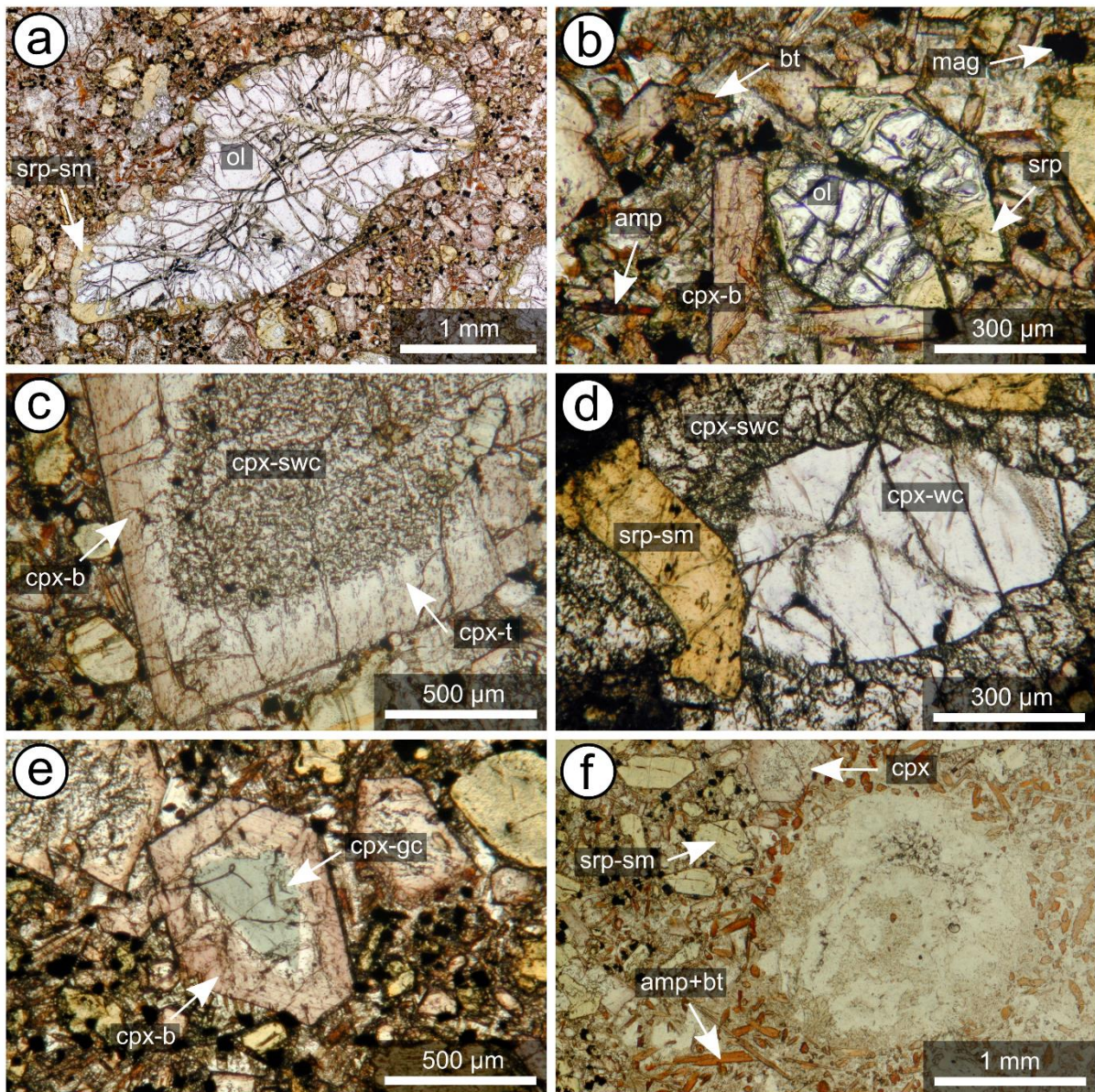


Fig. 4: Mineral textures in the monchiquites from Buggingen (transmitted light microscopy images). a) Large, rounded olivine (ol) macrocryst partly replaced by serpentine and smectite minerals. b) Euhedral olivine, brown clinopyroxene needles and small tabular amphibole (amp) and biotite constitute the groundmass. c) Clinopyroxene consisting of a sieved white core (cpx-swc), a plain transitional zone (cpx-t) and a brown rim (cpx-b). d) Clinopyroxene with an irregularly shaped white core (cpx-wc), surrounded by sieved white clinopyroxene and olivine that is completely replaced by serpentine and smectite minerals. e) Green-core clinopyroxene (cpx-gc) rimmed by brown clinopyroxene. f) Roundish zone consisting of a fine-grained mixture of chlorite, talc and calcite, surrounded by amphiboles and biotite.

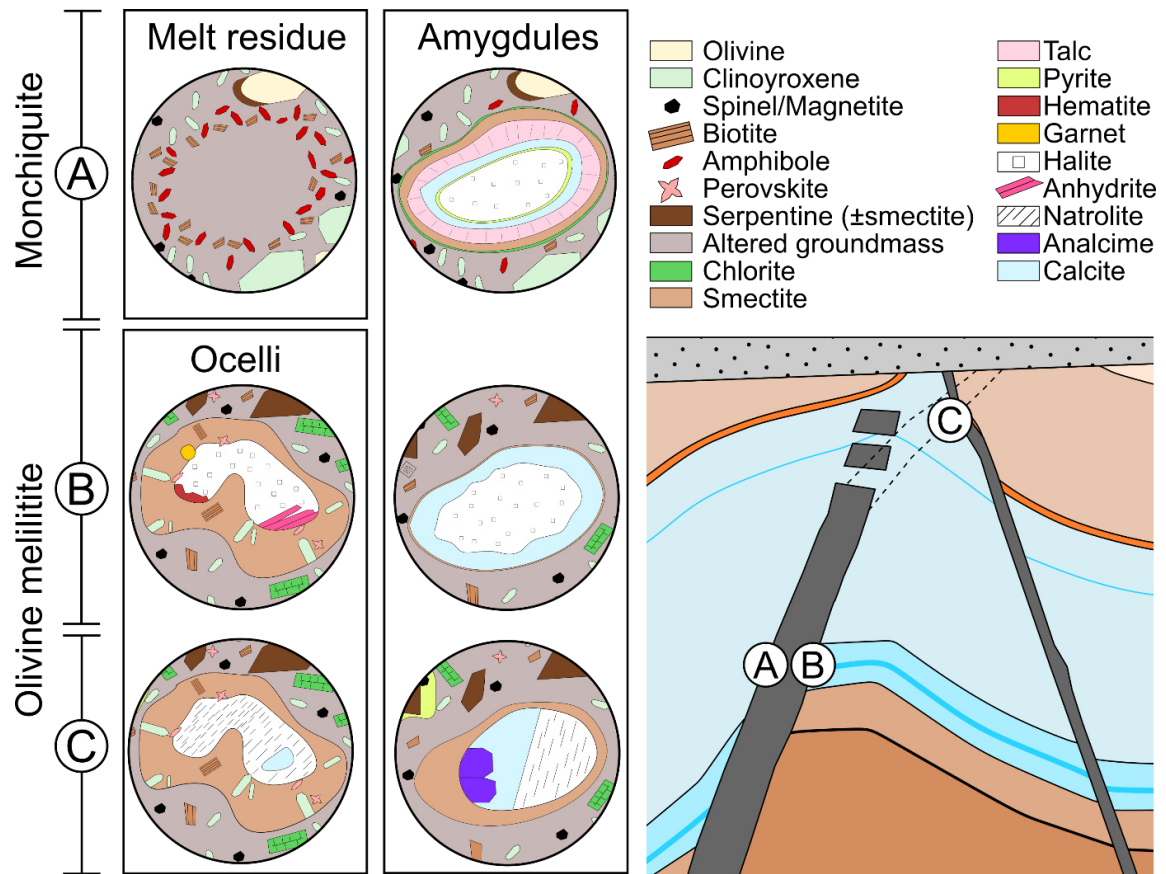


Fig. 5: Schematic overview of the spheroidal textures in the monchiquites and the olivine melilitites.

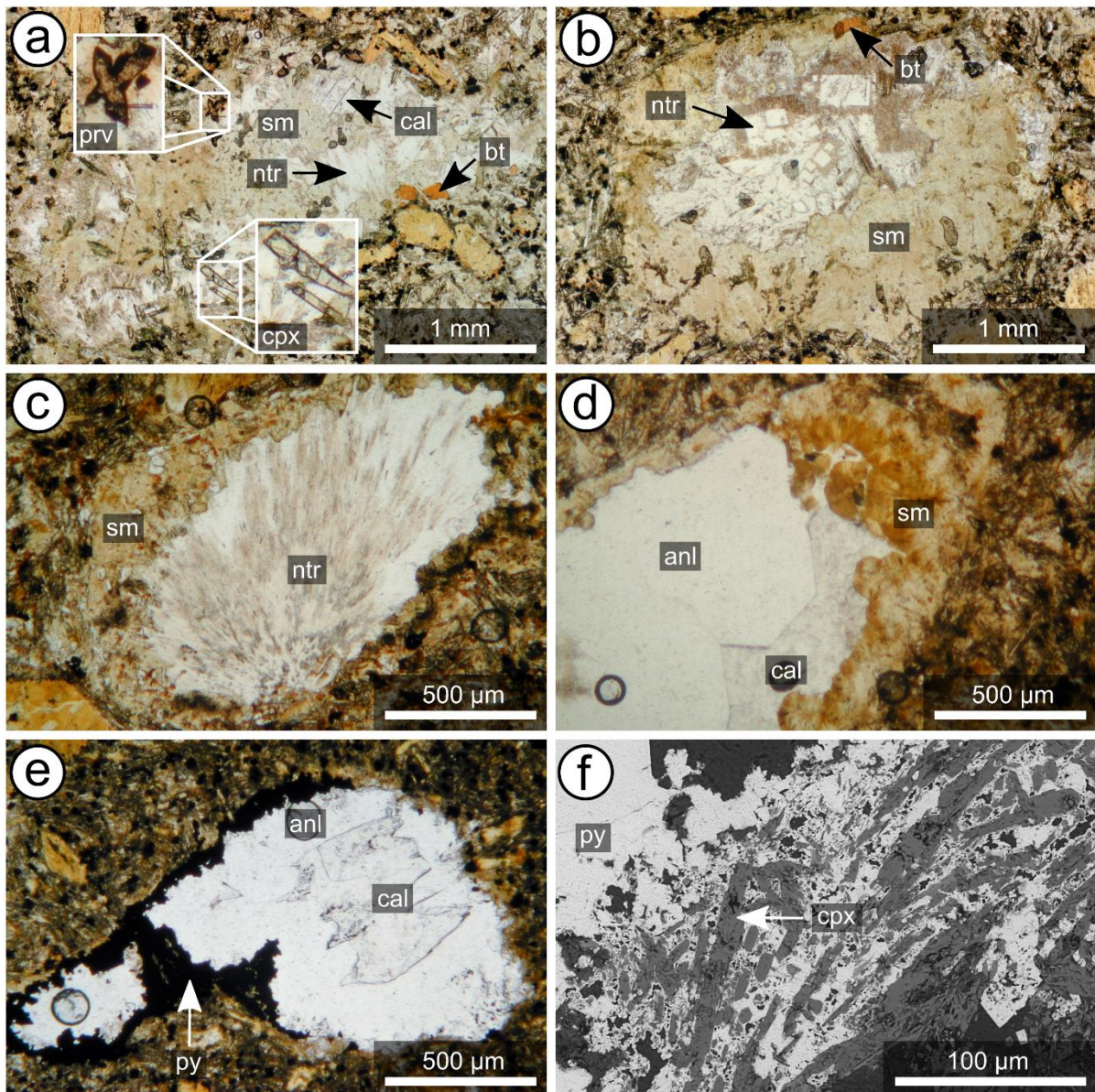


Fig. 6: Mineral textures of the ocelli in the olivine melilitites associated with the Fish Shale (transmitted light microscopy images. a) Biotite, clinopyroxene and perovskite (prv) sometimes occur at the rim of ocelli. Calcite (cal), smectite and natrolite are further common constituents. b) Ocellus with a rim mainly consisting of smectite and rare biotite, and a core of natrolite (ntr). c) Ocellus with marginal smectite minerals and a core of natrolite. d) Ocellus consisting of smectites, calcite and analcime (anl). e) Ocellus with marginal pyrite (py), analcime and calcite. f) Parts of the groundmass are replaced by pyrite.

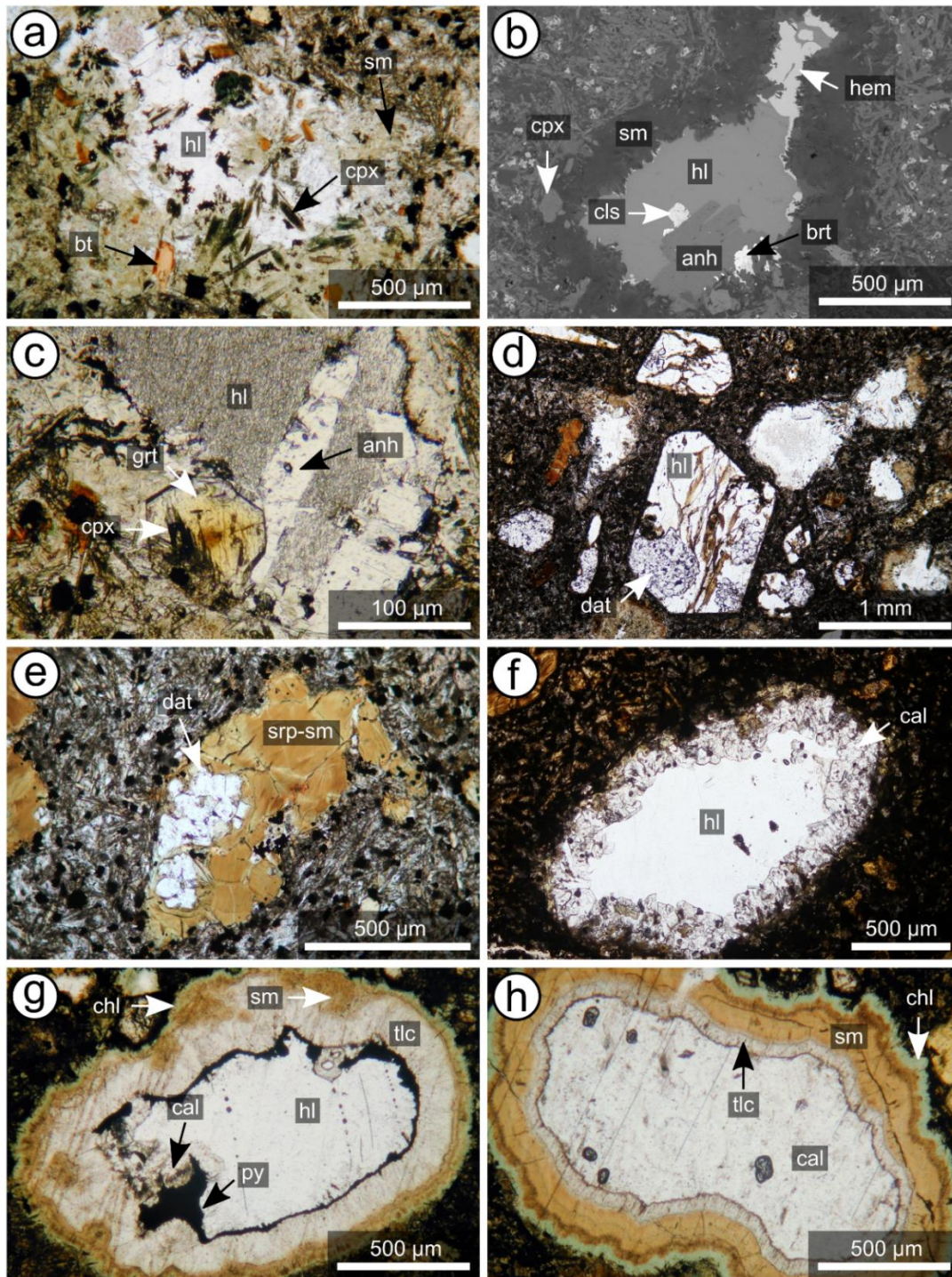


Fig. 7: Mineral textures of the ocelli and amygdules in the olivine melilitites and the monchiquites associated with the potash salt layer (transmitted light microscopy and BSE images [b]). a) Ocellus with a rim of biotite, aegirine and smectites, and a core of halite. b) Ocellus consisting of smectites, hematite (hem), barite (brt), anhydrite (anh), celestine (cls) and halite. c) In sample M156, euhedral garnet (grt) with inclusions of clinopyroxene sometimes occur at the rim of ocelli. d) Halite and datolite (dat) pseudomorph after olivine. e) Serpentine, smectite and datolite pseudomorph after olivine. f) Amygdule consisting of marginal calcite and a core of halite. g) Amygdule with a concentric succession of chlorite (chl), smectite, talc (tlc), calcite, pyrite (py) and halite. h) Amygdule showing a concentric succession of chlorite, smectite, talc and calcite.

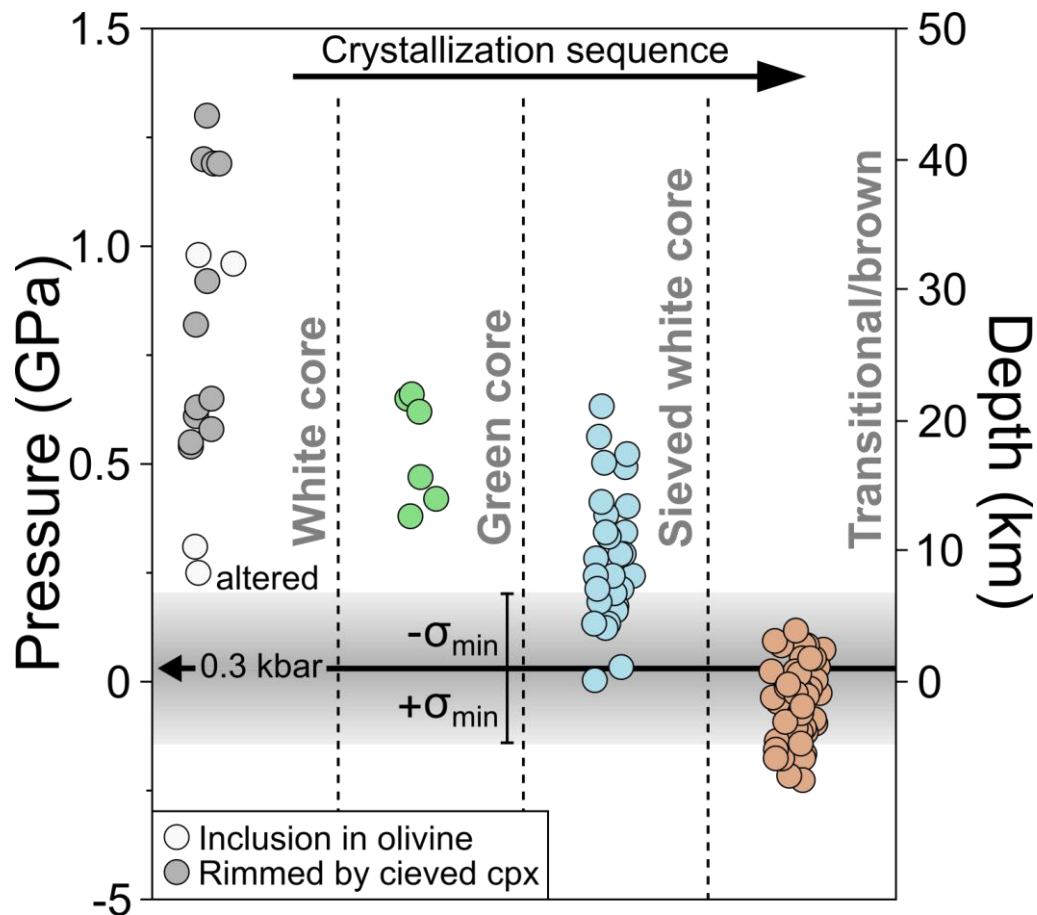


Fig.8: Crystallization depth for the various clinopyroxene types in the monchiquites from Buggingen. Pressure estimates are based on Nimis & Ulmer (1998) and indicate a minimum uncertainty of $\sigma=1.75$ kbar. The negative pressure estimates for many of the transitional/brown clinopyroxenes might additionally be caused by the fact that the barometer is calibrated for anhydrous systems (note that late-stage clinopyroxenes coexist with amphibole and mica). The line at 0.3 kbar demonstrates the expected minimum pressure during groundmass crystallization due to an overburden of ~ 1000 m.

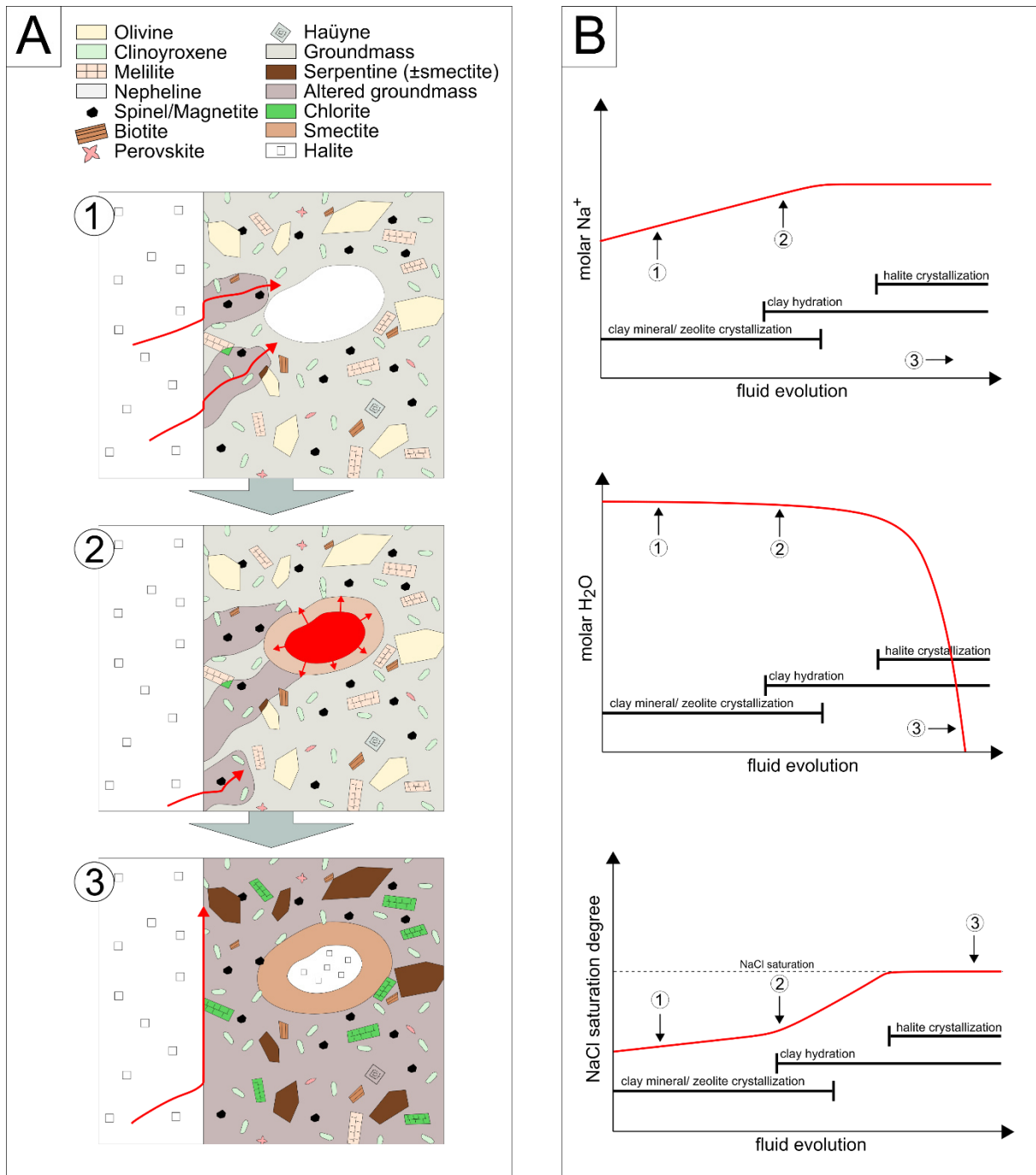


Fig. 9: Precipitation mechanism for the halite-bearing amygdules/ocelli. A) Schematic evolution of the dyke rock mineral assemblage. 1) NaCl-saturated fluid (red arrow) interacts with the groundmass, thereby producing Na- and OH-bearing alteration phases (smectites and zeolites). 2.) Fluid hits a vug, forms marginal smectite minerals and gets trapped. 3) Gradual removal of H₂O from the fluid by clay mineral swelling results in NaCl precipitation. The formation of smectites and zeolites cements possible fluid paths and prevents further fluids from entering the dyke rock (blue arrow). B) Schematic evolution of the fluid composition. The upper two xy-diagram demonstrate the evolution of the molar Na⁺, respectively H₂O concentration of the fluid with respect to the initial fluid composition. The lower xy-diagram demonstrates the NaCl saturation degree during fluid evolution.

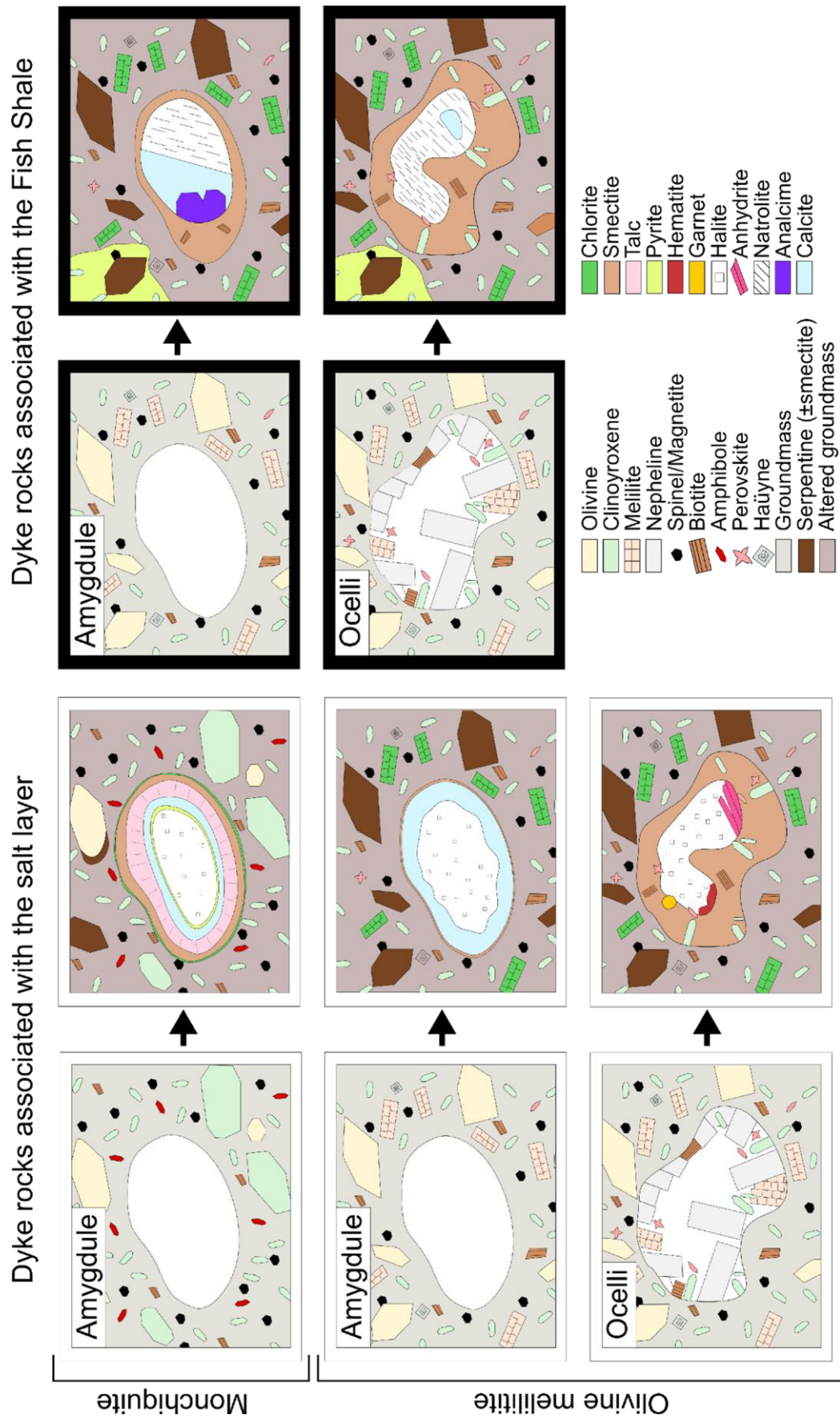


Fig. 10: Schematic diagram illustrating the mineral assemblage of amygdules and ocelli before and after the hydrothermal overprint.

Appendix VII

Accepted publication

Study 7

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Do carbonatites and alkaline rocks reflect variable redox conditions in their upper mantle source?

Simon Braunger^{a,*}, Michael A.W. Marks^a, Thomas Wenzel^a, Luanna Chmyz^b, Rogério Guitarrari Azzone^b, Gregor Markl^a

^a Department of Geosciences, Eberhard-Karls-Universität Tübingen, Wilhelmstrasse 56, 72074 Tübingen, Germany

^b Institute of Geosciences, Universidade de São Paulo, Rua do Lago 562, 05508-080 São Paulo, Brazil

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ABSTRACT

A detailed investigation on seven carbonatites and associated alkaline rock complexes (Kaiserstuhl, Sokli, Kovdor, Palabora, Oka, Magnet Cove, Jacupiranga), together with a world-wide comparison between carbonatites, alkaline silicate rocks and mantle xenoliths, implies peculiar redox conditions for carbonatite-bearing alkaline complexes: Carbonatites and associated alkaline rocks in continental settings crystallize from relatively oxidized magmas, on average 1.4 log units ($\sigma = 1.7$) and 1.3 log units ($\sigma = 1.1$) above the synthetic fayalite-magnetite-quartz (FMQ) buffer. In contrast, alkaline rocks in continental settings that lack associated carbonatites reveal rather reduced conditions (mean $\Delta\text{FMQ} = -1.1$; $\sigma = 1.4$). The calculated redox conditions for carbonatites and associated silicate rocks demonstrate that these crystallize from relatively oxidized mantle-derived melts compared to the general range found for alkaline rocks in continental settings.

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1. Introduction

Carbonatites are carbonate-rich magmatic rocks that typically occur in continental intraplate settings, such as at the edge of cratons or within rifts (Woolley and Kjarsgaard, 2008). Carbonatitic magmas can evolve from strongly SiO_2 -undersaturated and carbonate-bearing parental silicate melts by fractional crystallization and liquid immiscibility, with the latter being especially important in alkali-rich systems (e.g. Jones et al., 2013; Martin et al., 2013). They may also form directly by partial melting of carbonate-bearing peridotite (e.g. Beccaluva et al., 2017; Chmyz et al., 2019).

Carbonatites are typically not pure carbonate rocks but contain variable amounts of silicates (e.g. olivine, mica, garnet), oxides (e.g. magnetite, pyrochlore, perovskite), sulfides (e.g. pyrrhotite, galena, pyrite), and phosphates (particularly apatite). In a few cases, carbonatites are associated with apatite-magnetite-olivine±carbonate rocks (phoscorites) which are related to carbonatites by crystal fractionation or liquid immiscibility from a common parental magma (Krasnova et al., 2004b; Giebel et al., 2019a). However, most carbonatites are spatially and temporally associated with variable alkaline silicate rocks, which suggests a genetic link between both rock groups. The silicate rocks include ultramafic,

mafic, intermediate and evolved lithologies. Depending on the erosion level of the magmatic complex, their textures range from coarse-grained cumulitic (e.g., Palabora, Kovdor) to fine-grained and porphyritic (sub)volcanic textures (e.g., in the East Africa Rift); mixed examples exist as well (e.g., Magnet Cove, Kaiserstuhl). Based on the associated silicate rocks, carbonatites are divided into several genetic types, with nephelinite- and melilitite-clan carbonatites representing the most abundant ones. Other genetic types include for example kimberlite- and ailikite-clan carbonatites (Mitchell, 2005).

The source region for most carbonatites and associated alkaline silicate rocks is the metasomatized lithospheric and/or sublithospheric (asthenospheric) mantle (e.g. Bell and Simonetti, 2010; Braunger et al., 2018). However, carbonatites and spatially associated alkaline silicate rocks do not necessarily derive from the same magma source, as demonstrated by various isotope studies (Bell, 1998). In cases, carbonatites are cogenetic with only some of the exposed silicate rocks (Kaiserstuhl; Braunger et al., 2018) or even none of them (Jacupiranga; Beccaluva et al., 2017; Chmyz et al., 2019). The variable genetic relationships between carbonatites and spatially associated alkaline rocks reflect the heterogeneity of the metasomatized lithospheric mantle. Depending on their scale, these heterogeneities cause the formation of compositionally diverse melt volumes that may evolve and differentiate indepen-

* Corresponding author.

E-mail address: simonbraunger@hotmail.com (S. Braunger).

dently from each other or may mix with each other, eventually forming a carbonatite-alkaline silicate rock complex.

Depending on oxygen fugacity (f_{O_2}), carbon in the mantle is present either as reduced species (graphite, diamond, carbides and CH_4) or in its oxidized form (CO_2 and carbonates; Frost and McCammon, 2008; Dasgupta and Hirschmann, 2010; Stagno, 2019). Redox conditions of the (sub)lithospheric mantle vary by more than 8 log units around the synthetic fayalite-magnetite-quartz (FMQ) buffer (e.g. Foley, 2011). This heterogeneity is related to different tectonic settings: thick cratonic lithosphere is variably reduced (mean of $\Delta FMQ = -2.8$, being more reduced at greater depth) due to the pressure-dependent stability of Fe^{2+} - and/or Fe^{3+} -bearing minerals (e.g. Ballhaus and Frost, 1994). In contrast, the mantle above subduction zones shows high redox conditions (mean of $\Delta FMQ = +0.5$) because of the interaction with oxidizing melts from the subducted slab (Parkinson and Arculus, 1999). Digestion of crustal rocks by the mantle leads to further heterogeneity with respect to redox conditions over time (e.g. Foley, 2011; Rohrbach and Schmidt, 2011). Although the oxidation state of the upper mantle has been assumed to be stable over the past 3800 Ma (e.g. Canil, 2002; Li and Lee, 2004), recent studies suggest an increase of the redox conditions since the Archean (e.g. Gaillard et al., 2015; Aulbach and Stagno, 2016). For the asthenosphere, estimates based on investigations of abyssal (oceanic) peridotites or Mid Ocean Ridge basalts reveal average redox conditions slightly more reduced than for the lithosphere, possibly due to the loss of oxidative fluids and melts that infiltrated the overlying mantle (Frost and McCammon, 2008; Foley, 2011). Alternatively, the convecting upper mantle (at least to 150 km depth) might not follow the f_{O_2} -depth correlation mentioned above and, hence, is more oxidized than the continental lithosphere. This was recently suggested by Eguchi and Dasgupta (2018) for redox conditions of subducted lithologies, but it remains to be tested, if such assumptions can be extrapolated to asthenospheric peridotites.

Those alkaline silicate rocks that occur in similar settings as carbonatites (such as intraplate environments) show a similarly large spread in their redox conditions as the upper mantle, namely from $\Delta FMQ = -4$ to $+6$. This variability is qualitatively expressed by the mafic minerals in alkaline rocks: Clinopyroxene-garnet \pm titanite assemblages seem to be indicative for relatively oxidized crystallization conditions, whereas olivine-clinopyroxene-magnetite assemblages indicate variable, but mostly reduced conditions. In some extreme cases, rather unusual sodic clinopyroxene-amphibole-eudialyte \pm aenigmatite assemblages, so-called apaitic rocks, crystallize from some of the most reduced magmas known on Earth (Marks and Markl, 2017). Subduction zone-related lamprophyric lavas belong to the most oxidized silicate rocks known (Carmichael, 1991), but these are not considered further here because of their different tectonic setting.

Very little is known about redox conditions for carbonatitic magmas and even less so about redox conditions for carbonatites and their spatially associated alkaline silicate rocks. Here, we present the results of thermodynamic calculations for seven igneous complexes (Kaiserstuhl, Sokli, Kovdor, Palabora, Magnet Cove, Oka, Jacupiranga) that constrain redox conditions for carbonatites and associated alkaline silicate rocks. Our results, augmented by literature data, provide insight into the variable redox conditions under which alkaline silicate rocks may form and indicate that carbonatites (and their associated alkaline silicate rocks) crystallize from remarkably oxidized mantle-derived magmas, compared to alkaline rocks in general.

2. Geology and petrography

We investigated carbonatites, phoscorites and associated alkaline silicate rocks (ultramafic, melilititic and nephelinitic rocks)

that contain appropriate mineral assemblages for deducing their crystallization conditions (Table 1). We used well-known examples from Europe (Kaiserstuhl, Germany; Sokli, Finland; Kovdor, Russia), Africa (Palabora, South Africa), North America (Magnet Cove, USA; Oka, Canada) and South America (Jacupiranga, Brazil). These examples cover a range of ages (Proterozoic to Cenozoic) and emplacement depths (close to surface to >5 km) and are considered to represent typical carbonatite-alkaline silicate rock associations (Mitchell, 2005).

2.1. Kaiserstuhl (Germany)

The Kaiserstuhl volcanic complex (q.v. Braunger et al., 2018 and references therein) was active between 19 and 16 Ma and predominantly consists of porphyritic tephritic rocks, medium-grained sodalite monzogabbros (in the centre), and several phonolite domes that are all assumed to derive from a rather potassic basanitic parental magma. Minor melilititic, nephelinitic, and basanitic rocks represent a more sodic rock series, probably genetically related with the carbonatites of the Kaiserstuhl complex. The examined coarse-grained calcite carbonatites contain variable amounts of magnetite, euhedral but partly altered olivine or monticellite, apatite and small amounts of phlogopite and pyrochlore (Fig. 1a).

2.2. Sokli (Finland)

The Sokli alkaline complex (q.v. Lee et al., 2004 and references therein) is part of the Palaeozoic Kola Alkaline Province and intruded 370–360 Ma ago. It shows a concentric structure and is divided into an outer zone consisting of carbonate-rich metasomatic facies that replaced ultramafic rocks (mainly clinopyroxenites) and a magmatic core composed of three phoscorite-carbonatite stages and two subsequent dolomite carbonatite stages. The investigated phoscorites and carbonatites are characterized by variable amounts of olivine, magnetite, apatite, phlogopite, baddeleyite and carbonate. Phoscorite type 3 lacks olivine but contains individual ilmenite grains instead (Fig. 1b).

2.3. Kovdor (Russia)

The Devonian (~ 420 –370 Ma) Kovdor massif (q.v. Krasnova et al., 2004a and references therein) also belongs to the Palaeozoic Kola Alkaline Province and is a multiphase alkali-ultrabasic pluton with olivinites, clinopyroxenites, melilitolites and marginal ijolites to melteigites. The subsequently intruded phoscorite-carbonatite complex in the west of the massif includes five main phoscorite types (P1–P5) with the last three being associated with paired carbonatites (C1–C3). The phoscorites consist of variable amounts of olivine, magnetite and apatite while phlogopite, baddeleyite and carbonate represent minor mineral phases. Carbonatites have a similar mineral assemblage but are obviously carbonate-rich. Irregular ilmenite patches and variably shaped spinel grains (Fig. 1c) occur as exsolutions in magnetite while C2 carbonatites also contain ilmenite-magnetite assemblages of the composite type. The general grain size of the phoscorites and the magnetite content increases from P1 to P3 and from P4 to P5, while the olivine concentration decreases. Phoscorite P4 and its associated carbonatite contain tetraferriphlogopite instead of phlogopite and in C3 carbonatites, dolomite is the predominant carbonate phase.

2.4. Palabora (South Africa)

The 2060 Ma old Palabora carbonatite complex (q.v. Giebel et al., 2019a and references therein) is exposed as an elongated tripartite pipe-like body that predominantly consists of clinopyroxenites and small satellite bodies of syenites. The central Loolekop

Table 1
Mineral assemblages and intrinsic parameters for investigated rock types.

Locality	Rock type	Mineral assemblage	Temperature (°C)	ΔFMQ**	
Kaiserstuhl	Carbonatite	Cal+ap+ mag±ol±mtc+phl	840-900 ^{OM}	+1.8 ± 0.8 ^{OM}	
	Nephelinite-basanite	Cpx+nph+ol+mag+sp+pl±afs	990-1050 ^{B18}	+2.1 ± 0.9 ^{B18}	
	Melilite-häüynite	Sgm+ cpx+grt+mll +nph+ mag+prv +ap	950-1120 ^{B18}	+1.9 ± 3.4 ^{B18}	
	Melteigite	Nph+ cpx+grt+mag+pl+ttn+prv ±ap	~1000 ^E	+1.8 ± 0.7 ^{B18}	
	Sodalite Monzogabbros-Tephrites	Cpx+pl+mag±ol +sgm+ afs±amp±phl±ttn ±ap	880-1220 ^{B18}	+1.7 ± 1.3 ^{B18}	
	Phonolites	Afs +sgm+ grt+cpx ±nph± wo±pl±ttn±mag +ap	880-1130 ^{B18}	+1.2 ± 0.9 ^{B18}	
Sokli	Phoscorite 1	Cal+ mag+ol +ap+ phl +dol+bdy+ilm	530-770 ^{OM}		
	Carbonatite 2	Cal+ ol+mag+ilm +ap+ phl +dol+bdy	590-700 ^{OM}		
	Phoscorite 3	Dol+ mag+ilm +cal+ap+ phl +bdy	510-670 ^{IM}	0 ± 0.9 ^{IM}	
Kovdor	Phoscorite 1	Ol +ap+ mag+phl +cal±dol+bdy+ ilm	590-660 ^{IM} ; 440-790 ^{OM}	+2.3 ± 1.0 ^{IM}	
	Phoscorite 2	Mag +ap+ ol +cal±dol+ phl +bdy+ ilm	500-880 ^{OM}		
	Phoscorite 3	Mag +ap+cal+ ol±dol+phl +bdy+ ilm	590-840 ^{OM}		
	Carbonatite 1	Cal+ mag+ol+phl +ap±dol+bdy+ ilm	590-640 ^{IM} ; 550-630 ^{OM}	+1.9 ± 0.6 ^{IM}	
	Phoscorite 4	Mag +ap+ ol+phl +cal±dol+bdy+ ilm	550-590 ^{IM} ; 580-750 ^{OM}		
	Carbonatite 2	Cal+ mag±ol+phl +ap±dol+bdy+ ilm	560-620 ^{IM} ; 470-570 ^{OM}	+2.4 ± 0.4 ^{IM}	
	Phoscorite 5	Mag+ol+phl +dol+cal+ap+bdy+ ilm	400-850 ^{OM}		
Palabora	Carbonatite 3	Dol+cal+ mag+phl +ap+bdy+ ilm	520-540 ^{IM}		
	pair	Phoscorite	Mag+ol +cal+dol+ap+ phl +bdy+ ilm	610-650 ^{IM} ; 510-860 ^{OM}	+1.2 ± 0.8 ^{IM}
	Banded carbonatite	Cal+ Ol+mag +dol+ap+ phl +bdy+ ilm	530-580 ^{IM} ; 450-600 ^{OM}		
Magnet Cove	Transgressive carbonatite	Cal+ Ol+mag +dol+ap+ phl +bdy+ ilm	570-650 ^{IM} ; 510-760 ^{OM}	+0.9 ± 0.4 ^{IM}	
	Carbonatite	Cal+ap+ mtc+mag+prv+grt+sp+ilm	>850 SM	+0.1 ± 0.3 ^F	
Oka	Clinopyroxenite-ijolite	Cpx ±nph± grt+mag±ttn+prv +ap	~1000 ^E	+1.8 ± 1.5 ^F	
	Syenite	Afs+ cpx +pl+ mag+phl+ttn +ap	~900 ^E	+1.2 ± 0.7 ^F	
	Calcite-rich okaite	Mll +cal+ prv+mag+cpx+sp+ilm+phl +ap	620-680 ^{IM} ; >800 SM	+0.8 ± 0.9 ^{IM} ; +1.6 ± 0.4 ^F	
Jacupiranga	Carbonatite	Cal+ap+ phl+mag+ilm	580-660 ^{IM}	+1.3 ± 0.7 ^{IM}	
	Clinopyroxenite	Cpx+mag+ilm+prv +ap	~1000 ^E	+0.8 ± 2.3 ^F	
	Monzonite-syenite	Afs±pl+cpx+mag+ilm+phl+ttn±amp +ap	>800 ^A	+1.2 ± 0.9 ^F	

Note: Mineral phases are listed in descending order of frequency with the bold phases being used to calculate the crystallization conditions.

**Only redox estimates at $T \geq 600$ °C are listed because of unpredictable redox changes during further cooling (compare Fig. 3).

Abbreviations: afs = alkali feldspar; amp = amphibole; ap = apatite; bdy = baddeleyite; cal = calcite; cpx = clinopyroxene; dol = dolomite; grt = garnet; ilm = ilmenite; mag = magnetite; mll = melilite; mtc = monticellite; nph = nepheline; ol = olivine; (t)phl = (tetraferri)phlogopite; pl = plagioclase; prv = perovskite; sgm = sodalite-group minerals; sp = spinel; ttn = titanite; wo = wollastonite.

Crystallization conditions constrained using: ^A Amphibole; ^{B18} Braunger et al., 2018; ^E Estimation; ^{IM} Ilmenite-magnetite; ^{OM} Olivine-magnetite. ^F Further mineral phase equilibria. See text for further information.

pipe additionally contains phoscorites and associated banded carbonatites that are partly penetrated by a transgressive carbonatite. The mineralogy of the different phoscorites and carbonatites is very similar but with variable amounts of olivine, magnetite (Fig. 1d), apatite, phlogopite, baddeleyite and carbonate. The core of magnetite commonly contains ilmenite and spinel exsolutions, while ilmenite-magnetite of the composite type is typical for the banded carbonatites.

2.5. Oka (Canada)

The 126-115 Ma old Oka carbonatite complex (q.v. Chen and Simonetti, 2013 and references therein) comprises two intrusive centres with an early formed ring of alternating carbonatites and silicate rocks and a later central carbonatite plug. The silicate rocks predominantly consist of ijolites and okaites with the latter only occurring in the northern ring and at least partly representing a transition to carbonatites. The calcite-rich okaite (up to 20% carbonate) consists of euhedral melilite, perovskite and clinopyroxene, euhedral to anhedral biotite and small amounts of apatite (Fig. 1e). Variably sized and shaped spinel exsolutions and rare ilmenite exsolutions are present within the subhedral magnetite grains.

2.6. Magnet Cove (USA)

The mid-Cretaceous (100±5 Ma) Magnet Cove alkaline complex (e.g. Eby, 1987) is an elliptically shaped ring-dyke consisting of

marginal foid syenites, clinopyroxenites (north-western and north-eastern edge of the complex), trachytes and phonolites that crop out as an intermediate ring, and central ijolites which were intruded by several carbonatite bodies. The studied carbonatites contain large columnar grains of apatite, subhedral monticellite, and magnetite whose core commonly consists of spinel and sometimes ilmenite exsolutions. Perovskite forms euhedral to subhedral crystals of various sizes and is occasionally associated with anhedral garnet. Large calcite crystals surround all previously mentioned mineral phases (Fig. 1f).

2.7. Jacupiranga (Brazil)

The early Cretaceous (135-131 Ma) Jacupiranga carbonatite complex (q.v. Chmyz et al., 2017 and references therein) can be divided into two silicate intrusions: 1) the northern part comprises peridotites (dunites and scarce wehrlites) that are surrounded by a heterogeneous zone of primitive to evolved rock types (e.g. lamprophyres, diorites, monzonites, syenites) while 2) a clinopyroxenite plug, with some occurrence of ijolites, melteigites and 5 pipe-like carbonatite bodies are present in the south. Besides calcite, the main mineral phases in the investigated carbonatites are apatite, phlogopite and magnetite. Ilmenite is present as individual grains but also surrounds magnetite and occurs as exsolutions in the latter (Fig. 1g).

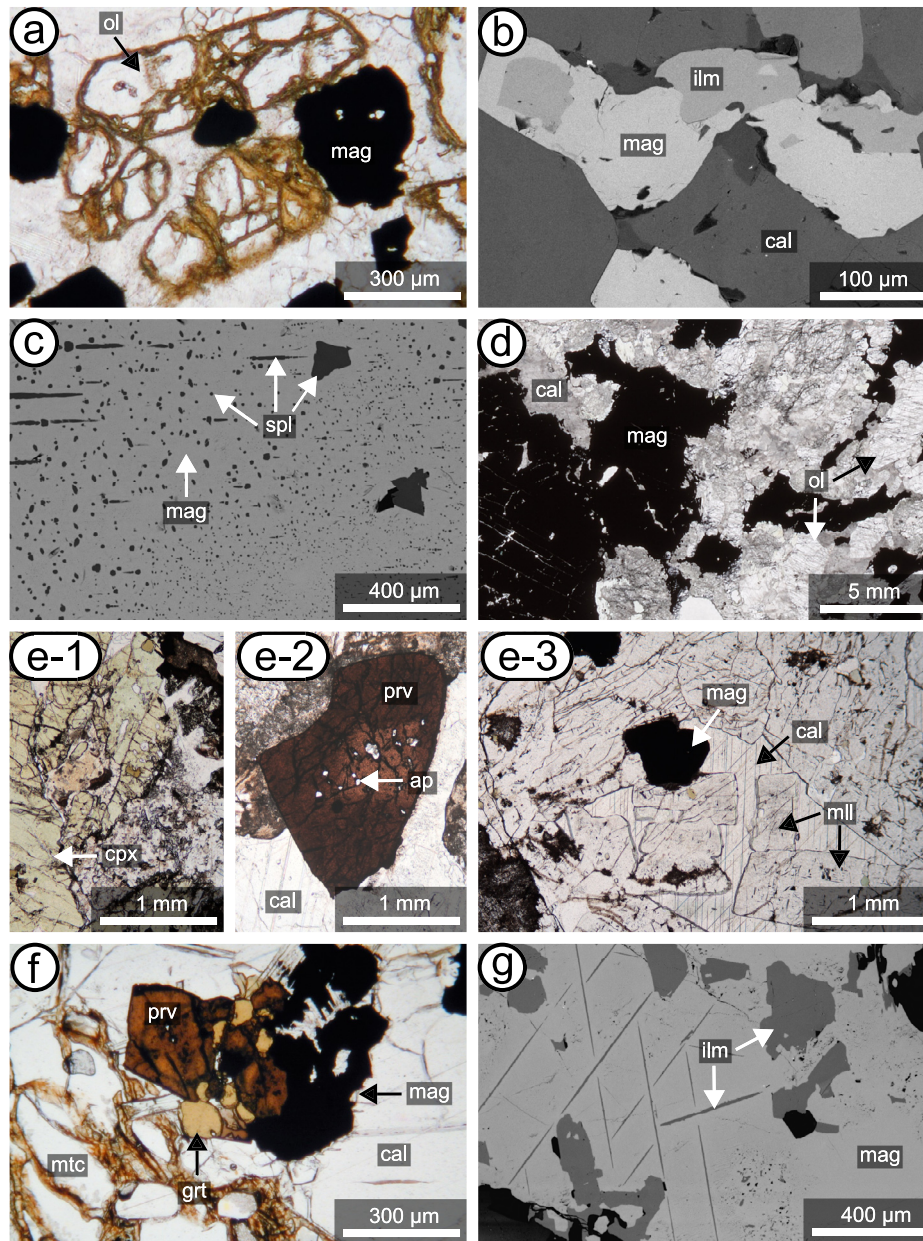


Fig. 1. Mineral textures in carbonatitic rocks (transmitted light microscopy [a, d, e, f] and BSE images). (a) Carbonatite from Kaiserstuhl consisting of partly altered olivine and magnetite. (b) Carbonatite from Sokli with an ilmenite-magnetite assemblage of composite type. (c) Phoscorite from Kovdor with elongated, rounded and irregularly shaped spinel exsolutions in magnetite. (d) Phoscorite from Palabora with large magnetite, olivine and calcite. (e) Okaite from Oka containing clinopyroxene (e-1), perovskite (e-2), melilite, magnetite and calcite (e-3). (f) Carbonatite from Magnet Cove consisting of perovskite (with apatite inclusions), garnet, magnetite and monticellite. (g) Carbonatite from Jacupiranga showing trellis-type and irregularly shaped ilmenite exsolutions in magnetite.

3. Methods

3.1. Analytical methods

Mineral compositions were analyzed using a JEOL JXA8900 electron microprobe in wave-length dispersive mode at the Department of Geosciences, University of Tübingen (Germany). An acceleration voltage of 20 kV and a beam current of 12 nA for garnet, perovskite and titanite, respectively 20 nA for all other mineral analyses was chosen. Peak counting times for major elements were 16 s and between 30 and 60 s for minor elements while the background counting times were half as long. Natural and synthetic standards were used for calibration. Peak overlaps between Ba-Ti, Ti-Ba, V-Ti, V-Cr, Al-Ba, F-Fe, Sm-Ce and Pr-La were corrected and an internal ZAF (spinel group minerals, ilmenite, garnet) or $\phi\rho z$

correction was applied for the raw data (Armstrong, 1991). Measurements were executed with a focused beam for olivine group minerals, pyroxene, melilite, spinel group minerals and ilmenite, but was widened to 2 μm for amphibole. In order to avoid a destruction of the grains due to the long measurement time for garnet, perovskite and titanite, the beam spot was set to 5 μm for these minerals.

3.2. Quantification of intrinsic parameters

Temperature and oxygen fugacity (f_{O_2}) during the crystallization of carbonatites and associated alkaline silicate rocks were calculated by applying several thermometers (magnetite-spinel exsolution, olivine-spinel, amphibole; Lehmann and Roux, 1986; Andersen et al., 1993; Ridolfi and Renzulli, 2012) and thermo-

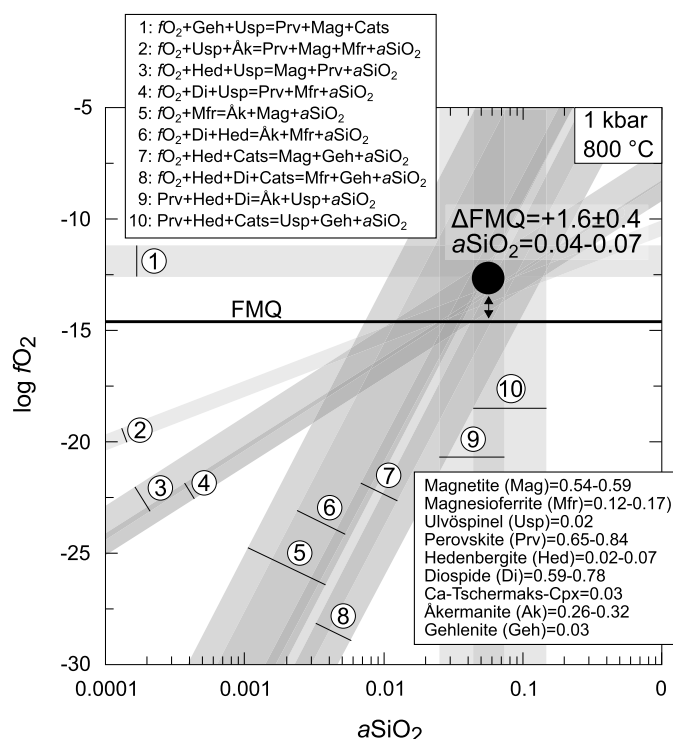


Fig. 2. Estimated oxygen fugacity and silica activity for the okaite from Oka, based on coexisting mineral assemblages. Reactions were performed with activity-corrected endmember compositions at a temperature of 800–900 °C and a pressure of 1 kbar. The black dot overlaps with all mineral equilibria and is the best fit for the redox conditions ($\Delta\text{FMQ} = +1.1\text{--}2.5$) and the silica activity ($a\text{SiO}_2 = 0.04\text{--}0.07$).

oxybarometers (ilmenite-magnetite; Andersen and Lindsley, 1985), as well as the *Perple_X* software package (Connolly, 1990) for computing activity-corrected phase diagrams (cf. Braunger et al., 2018 for the applied mineral endmember solution models) using appropriate mineral assemblages and textures. The implemented thermodynamic database of Holland and Powell (1998) was extended by thermodynamic properties for perovskite (Robie and Hemingway, 1995) and was checked for internal consistency. Pressure estimates were taken from literature data, except for Jacupiranga where they are based on amphibole barometry (Ridolfi and Renzulli, 2012) in associated monzonites.

The calculated intrinsic parameters were then augmented by literature data on crystallization conditions of carbonatites and associated silicate rocks. In order to avoid over-representation by multiple redox determinations for the same or similar rock types within a complex, the different rock types were summarized into subgroups according to their state of evolution. A *t*-test was performed to check the statistical significance of the differences between carbonatites, respectively associated silicate rocks and alkaline silicate rocks without association to carbonatites with regard to redox conditions (see the electronic supplement for a complete compilation of the data and methods used in this study).

4. Results

High temperatures (typically above 750 °C) were obtained for phoscorites from Sokli, Kovdor and Palabora, and for carbonatites from the Kaiserstuhl using core analyses of olivine and magnetite. In cases where magnetite cores show exsolution textures, the primary chemical composition was reintegrated. Associated carbonatites at Sokli, Kovdor and Palabora indicate lower maximum temperatures (570–760 °C), while calculations using magnetite and olivine rim analyses generally yield >200 °C lower temperatures. The carbonatitic rocks from Magnet Cove and Oka also yield high temperatures (>800 °C) based on spinel exolutions in magnetite.

Temperatures for carbonatites from Jacupiranga were estimated to ≤ 660 °C (ilmenite-magnetite equilibria). For the other carbonatites and phoscorites, ilmenite-magnetite assemblages yield temperatures in the range of 530 to 680 °C. Silicate rocks reveal temperatures typically above 800 °C (e.g. amphibole thermometry for monzonites at Jacupiranga), although some temperatures are rough estimates only (e.g. clinopyroxenites at Magnet Cove and Jacupiranga; Table 1).

Redox conditions for the carbonatites from Kaiserstuhl ($\Delta\text{FMQ} = +1.8 \pm 0.8$) were calculated using coexisting olivine-magnetite pairs assuming a silica activity slightly below the diopside stability (Fig. 6 in Braunger et al., 2018), since the latter is only present where the carbonatitic magma interacted with the silicate wall-rock, leading to a raised silica activity (see details in Giebel et al., 2019b). Redox conditions for carbonatites and phoscorites from Sokli ($\Delta\text{FMQ} = 0 \pm 0.9$), Kovdor ($\Delta\text{FMQ} = +2.2 \pm 1.0$) and Palabora ($\Delta\text{FMQ} = +1.2 \pm 0.8$), as well as for the okaite from Oka ($\Delta\text{FMQ} = +0.8 \pm 0.9$) and the carbonatites from Jacupiranga ($\Delta\text{FMQ} = +1.3 \pm 0.7$) were determined by ilmenite-magnetite assemblages. More complex mineral assemblages including minerals like clinopyroxene, melilite, magnetite, garnet and perovskite (Fig. 2) constrain the redox conditions for the carbonatites from Magnet Cove ($\Delta\text{FMQ} = +0.1 \pm 0.3$), the okaite from Oka ($\Delta\text{FMQ} = +1.6 \pm 0.4$), and for the alkaline silicate rocks at Kaiserstuhl ($\Delta\text{FMQ} = +1.9 \pm 3.4$), Magnet Cove ($\Delta\text{FMQ} = +1.8 \pm 1.5$) and Jacupiranga ($\Delta\text{FMQ} = +0.8 \pm 2.3$).

5. Discussion

5.1. Crystallization interval of carbonatites

The temperature interval recorded in various carbonatites by variable techniques is very broad (Fig. 3a). The highest (probably near-liquidus) temperatures of up to about 1000 °C are indicated by homogenization temperatures for melt inclusions in various

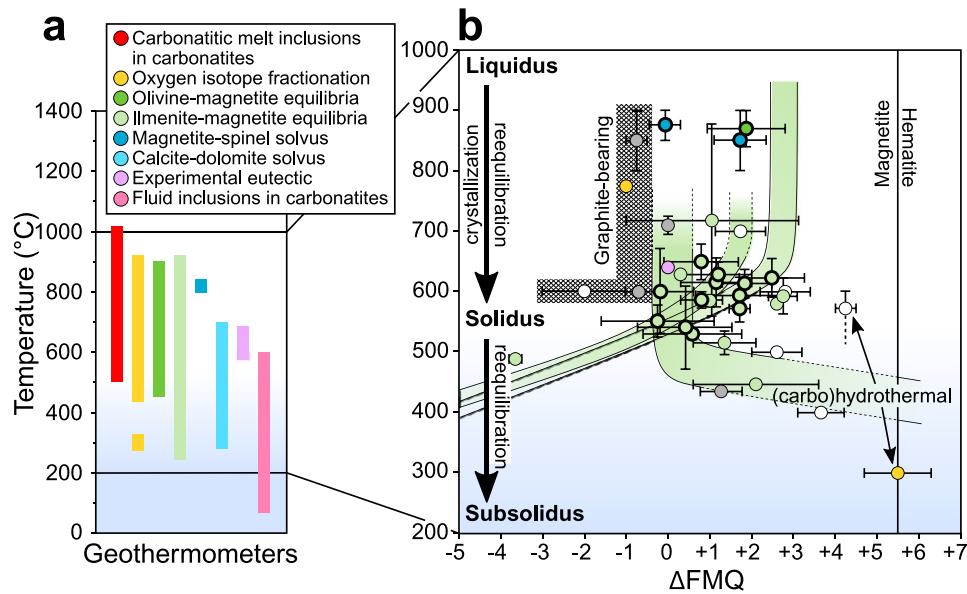


Fig. 3. Crystallization conditions for carbonatitic rocks. (a) Temperature interval constrained by several geothermometers. (b) Redox conditions in dependence of temperature. The color code of the dots refers to the applied geothermometers of Fig. 3a, grey dots represent further temperature determinations, white dots are rough estimates, bold-rimmed data points are new calculations of this study. Each data point refers to a specific redox calculation method applied to a particular carbonatitic rock type of a locality. Green paths indicate redox-temperature changes based on ilmenite-magnetite equilibria (extrapolated parts are highlighted by dashed lines) for carbonatites from Palabora, Kovdor (low redox conditions at low temperature) and Newania. Carbo-(hydro)thermal processes presumably result in an increase of the redox conditions, while dry thermal reequilibration lead to a decrease of ΔFMQ (see the electronic supplement for references of the literature data). (For interpretation of the colors in the figure(s), the reader is referred to the web version of this article.)

early magmatic minerals (such as olivine, monticellite, clinopyroxene and magnetite; (e.g. Guzmics et al., 2011). Oxygen isotope fractionation between apatite, magnetite, olivine and phlogopite (e.g. Doroshkevich et al., 2010) as well as magnetite-ilmenite equilibria preserve temperatures of up to around 900 °C (e.g. at Kaiserstuhl, Sokli, Kovdor, Palabora). The spread to much lower apparent temperatures (down to about 300 °C) is caused by variable reequilibration processes during cooling. For example, the apparently lower temperatures for carbonatites compared to their associated phoscorites (see above) are probably related to the smaller grain size of olivine and magnetite, which facilitates intense syn- to postmagmatic diffusion (e.g. Ozawa, 1984). Similar effects probably cause the overlapping temperature interval between ilmenite exsolutions in magnetite (synmagmatic or subsolidus?) and presumably magmatic composite ilmenite-magnetite assemblages (Fig. 3a). Near-solidus temperatures (≤ 700 °C) are recorded by e.g. dolomite exsolutions in calcite (Zaitsev and Polezhaeva, 1994) and experimental data constrain the eutectic of simple carbonatitic compositions to ~ 650 °C (Lee et al., 2000). The transition from magmatic to (carbo)hydrothermal conditions is determined by fluid inclusion homogenization temperatures (up to 600 °C; Rankin, 1975) and even lower temperatures (down to 530 °C) are indicated by melt inclusions trapped in late-stage carbonate that roughly mark the carbonatite solidus (e.g. Sokolov, 2014). In summary, carbonatites record a liquidus-solidus interval from ~ 1000 °C to roughly 500 °C (depending on composition). In the following only redox estimates calculated at temperatures above 600 °C are taken into consideration in order to minimize potential fractional crystallization and thermal reequilibration issues (Fig. 3b).

5.2. Redox conditions during the crystallization of alkaline rocks and carbonatites

On average, alkaline silicate rocks that are not associated with carbonatites formed under statistically significant ($p < 0.001$) more reduced conditions (mean $\Delta\text{FMQ} = -1.1$, $\sigma = 1.4$; Fig. 4a) than alkaline silicate rocks associated with carbonatites (mean

$\Delta\text{FMQ} = +1.4$, $\sigma = 1.7$; Fig. 4b) and carbonatites themselves (mean $\Delta\text{FMQ} = +1.3$, $\sigma = 1.1$; Fig. 4c). In our compilation, we only included alkaline silicate rocks from continental settings since carbonatites mainly occur in such geodynamic environments. Carbonatites at present subduction zones are so far only known from Mt. Vulture (Italy), but it is debated that carbonatite genesis at this locality is not related to the subduction process itself (cf. D’Orazio et al., 2007; Stoppa et al., 2008). We also excluded very rare carbonatites from ocean islands reported only from the Canary Islands, Cape Verdes and Kerguelen (Woolley and Kjarsgaard, 2008) and rare graphite- or diamond-bearing carbonatites (e.g. Chernigovsky, Chagatai, Yonghwa) as these constitute only $<2\%$ of all carbonatite complexes worldwide and would therefore be over-represented in our compilation (they would constitute 30% of the carbonatite complexes for which redox conditions are available; Fig. 4c). Group I kimberlites are genetically closely related to carbonatites as they also occur in intraplate settings (mainly within cratons), require a carbonate-rich mantle source, are characterized by extreme levels of silica undersaturation and may contain significant amounts of carbonate (Mitchell, 1986). Kimberlites have been frequently studied but, mostly because of their hybrid nature (Russell et al., 2013) and CO_2 degassing processes (Dongre and Tappe, 2019), their redox conditions vary by more than 16 log units (see electronic supplement Fig. 1). We therefore chose not to integrate them into our compilation and will not further discuss the kimberlite group.

5.3. Redox conditions of the source region of alkaline rocks and carbonatites

It is widely assumed that primitive mantle-derived rocks inherit the oxidation state of their source region (e.g. Carmichael, 1991), although some variance in the redox estimates may result from fractional crystallization of Fe-bearing minerals and degassing processes (Gaillard et al., 2015). Although evolved rocks are generally more susceptible to such effects, a very similar range of redox conditions compared to the primitive rocks exists (Fig. 4a&b). The

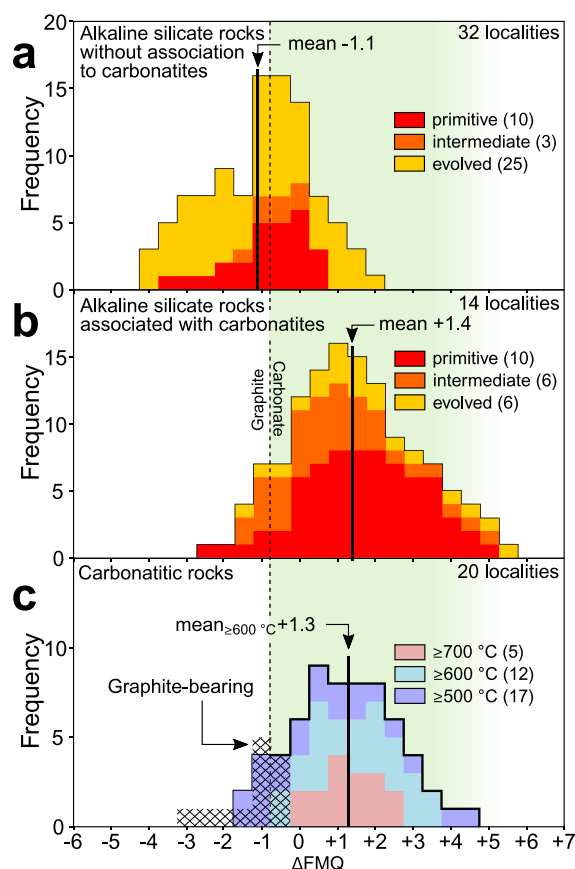


Fig. 4. Histograms compiling redox conditions for (a) alkaline silicate rocks without association to carbonatites, (b) those being related to carbonatites and (c) carbonatitic rocks. Alkaline silicate rocks are distinguished into primitive, intermediate and evolved types, carbonatites are sub-classified based on temperature. Numbers in brackets indicate the sample size for each group. The vertical dashed line in all three subfigures shows the stability field of graphite-, respectively carbonate-bearing peridotite.

good accordance between carbonatites and (primitive) associated silicate rocks indicates that the former may also reflect the redox state of their mantle source. However, during partial melting of a mantle rock, ferric iron preferably partitions into the melt while ferrous iron remains in its residue (e.g. Canil and O'Neill, 1996). Hence, the melt tends to record higher redox conditions than the source rock prior and after melt extraction. In contrast, the f_{O_2} -P dependency of a basaltic melt in a closed system differs from that of solid mineral assemblages leading to a relatively more reduced oxidation state of the melt at crustal conditions compared to the oxidation state of the mantle source. Moreover, the f_{O_2} -P path of synthetic buffers like FMQ deviates from those of basaltic melts and solid mineral assemblages which further complicates the comparison of ΔFMQ at mantle pressure with that at crustal levels (e.g. O'Neill et al., 2006). The interplay of all these factors has not yet been sufficiently investigated and further studies in this regard are warranted. We assume that the complications mentioned above should on average have similar affects for alkaline complexes with and without associated carbonatites. Therefore, we suggest that differences in the calculated oxygen fugacity of igneous rocks are an effect of variable redox conditions of the mantle.

Support for relatively oxidized conditions in the source region of carbonatites comes from redox estimates on mantle xenoliths that show trace element signatures characteristic for the interaction between mantle and carbonate-rich liquids. These mantle fragments are up to 4 log units more oxidized than typical lithospheric mantle (Fig. 5) and were found in various geodynamic environments, such as orogenic massifs (Pyrenees; Woodland et al., 1996), cratons (Slave, Kaapvaal; Creighton et al., 2010; Hanger et al., 2015) and other intraplate settings (Massif Central; e.g. Uenver-

Thiele et al., 2017) including continental rift environments (East African rift; Rudnick et al., 1994). The Massif Central and particularly the East African Rift also comprise several carbonatite occurrences. We suggest that such xenoliths not only reflect interaction between lithospheric mantle and (asthenosphere-derived) carbonate-rich liquids, but that the xenoliths derive from a mantle that may reflect one possible source for carbonatite-bearing complexes.

Also thermodynamic calculations indicate that the redox conditions for carbonatites and associated alkaline rocks might reflect the oxidation state of their mantle source. The minimum redox conditions for the melting of carbonated peridotite are constrained by the enstatite-magnesite-olivine-diamond/graphite buffer reaction, which marks the stability limit for carbonates in peridotite (EMOD/G in Fig. 5) for pressures above 2 GPa. At lower pressures, CO_2 is stable instead of carbonate and the minimum redox conditions are limited by the graphite- CO_2 buffer (D/GCO₂ in Fig. 5). The most oxidized mantle rocks known (Foley, 2011) define an empirical upper f_{O_2} which strongly depends on pressure ($-0.7\Delta\text{FMQ}/\text{GPa}$; Fig. 5) due to the stability shift towards Fe^{2+} instead of Fe^{3+} -bearing mineral phases with depth (e.g. Ballhaus and Frost, 1994). Note that the upper f_{O_2} limit for carbonated peridotite and the f_{O_2} -depth correlation of the mantle is solely based on data from lithospheric xenoliths, but we suggest that the asthenospheric mantle should behave similarly because of its similar mineralogy.

Carbonatites often occur in intraplate-rift settings or at the thinned edge of cratons, consistent with the formation of nephelinitic to melilititic melts at mantle depths between 50-130 km (Hirose and Kushiro, 1993; Gudfinnsson and Presnall, 2005). Based

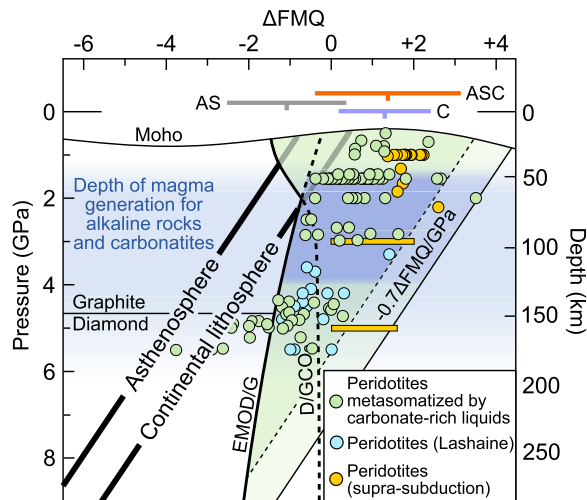


Fig. 5. Pressure and redox conditions for the solidus of carbonated peridotite limited by the enstatite-magnesite-olivine-graphite/diamond (EMOD/G) buffer, the diamond/graphite- CO_2 (D/GCO₂) equilibrium and an empirically constrained boundary (slope of $-0.7\Delta\text{FMQ}/\text{GPa}$; shift towards lower ΔFMQ by excluding outliers; Ballhaus and Frost, 1994). At pressures <2 GPa, peridotite is in equilibrium with CO_2 instead of carbonate. Mean redox estimates (with standard deviation) of (AS) alkaline silicate rocks without association to carbonatites, (ASC) alkaline silicate rocks associated with carbonatites and (C) carbonatitic rocks are shown for comparison. Peridotite xenoliths that occur in carbonatitic tuffs from Lashaine (Tanzania) are presumably metasomatized by a silica-rich melt and are not directly related to carbonatite generation (Rudnick et al., 1994; Bell and Simonetti, 2010).

on experiments (e.g. Wyllie et al., 1990), the parental melts for carbonatites originate from depths >75 km (>2 GPa). The depth interval of 75–130 km constrains the redox conditions of the source region of most carbonatitic melts to $\Delta\text{FMQ} = -1$ to $+3$ (Fig. 5), which also agrees with the oxidation state of carbonatites and their associated alkaline silicate rocks (Fig. 4b&c). The lower f_{O_2} conditions determined for alkaline complexes that lack carbonatites imply an ordinary or even reduced mantle source (Fig. 5). However, most carbonatites derive from oxidized magmas at rather shallow depth, since the stability field of carbonate-bearing peridotite continuously diminishes with depth. Furthermore, the redox state of the mantle decreases with depth, which makes it more difficult to oxidize graphite/diamond to carbonate (Fig. 5). Indeed, carbonatites that are cogenetic with lamprophyres and kimberlites may originate from greater depths (>150 km; e.g. Tappe et al., 2017), but these account for only a minor fraction of the known carbonatite occurrences (Woolley and Kjarsgaard, 2008).

6. Implications and conclusions

Although carbonatites and spatially associated alkaline silicate rocks do not always derive from the same magma source, carbonatites and associated silicate rocks in the 7 investigated complexes all indicate oxidized conditions (Table 1), suggesting that mantle oxidation processes must have affected all tapped mantle domains. Similarly, our world-wide compilation demonstrates that carbonatites and alkaline silicate rocks associated with carbonatites formed under rather oxidized conditions compared to the general range found for alkaline rocks in intraplate settings.

Especially the average asthenosphere, except for the uppermost part, is presumably too reduced to stabilize CO_2 or carbonate (Fig. 5, Foley, 2011). However, since partial melting processes within the mantle strongly affect its oxygen fugacity (Canil and O'Neill, 1996), there still may be domains oxidized enough for the formation of carbonates, and hence for carbonate-rich melts. A relatively high oxidation state of the mantle can also be achieved by the digestion of subducted crustal material (e.g. Yaxley and Brey, 2004; Rohrbach and Schmidt, 2011), the influence of which has been revealed for many young (<300 Ma) carbonatites by boron isotopes (Hulett et al., 2016). Carbonate-bearing sediments and altered oceanic crust, which contains around 2.5 wt.% CO_2 in the top

few hundred meters (Alt and Teagle, 1999; Yaxley et al., 2019), additionally raises the carbonate concentration of the mantle and further promotes the formation of carbonate-rich melts. (note that altered oceanic crust reveals carbon isotope systematics very similar to the average mantle and carbonatites; Deines, 2002; Shilobreeva et al., 2011). Such C-enriched and oxidized asthenospheric mantle domains may be the ultimate source for carbonatite complexes, or they function as an intermediate stage by providing carbonate-rich melts that infiltrate the base of the lithosphere, thereby finally preconditioning the source region for carbonatites and associated silicate rocks (Foley and Fischer, 2017).

An oxidized mantle with a sufficiently high carbonate content is not the only prerequisite for the formation of carbonatite complexes. The mantle beneath subduction zones is generally oxidized enough to stabilize carbonate (Foley, 2011) but carbonatites directly related to subduction processes seem to be lacking and Yaxley et al. (2019) argue that this is because of the reversed temperature gradient of the mantle wedge above the subducting plate. Potential carbonate-bearing melts would react with the overlying mantle, thereby being strongly modified and diluted. Ocean island volcanics and xenoliths from oceanic environments also indicate elevated oxygen fugacities (mean $\Delta\text{FMQ} \approx +0.6$; Ballhaus, 1993; Mallmann and O'Neill, 2007; Foley, 2011). Moreover, ocean island basalts reveal similar radiogenic isotope signatures as carbonatites and are suggested to form by the involvement of recycled oceanic crust (e.g. Hauri et al., 1993; Bell and Simonetti, 2010). The absence of carbonatites at most ocean islands is therefore not a result of too reduced conditions, but of a too thin oceanic lithosphere leading to primitive melt compositions that will not evolve into the carbonatite-silicate miscibility gap. Exceptions are the carbonatite occurrences at the Cape Verdes and Canary Islands, which are situated on a much thicker oceanic lithosphere and whose primitive silicate melts have an extraordinary low silica and high alkali content, allowing the formation of carbonatites during fractionation (e.g. Martins et al., 2010; Martinez-Arevalo et al., 2013; Weidendorfer et al., 2016; Schmidt and Weidendorfer, 2018).

Because of the known heterogeneity in the source region, the absence of carbonatites in some oxidized alkaline silicate complexes (e.g. Katzenbuckel, Ankaratra, Itatiaia; Mann et al., 2006; Cucciniello et al., 2017; Melluso et al., 2017) may therefore reflect a lack of carbon in the mantle source, crystallization of the

silicate melt before an exsolved/residual carbonatitic magma could form, decarbonation reactions in the melt which released CO₂-rich fluids, or simply lack of outcrop. Likewise, silicate rocks associated with carbonatites may form under reduced conditions as well, as suggested by existing examples of olivine-clinopyroxene-magnetite syenites associated with carbonatites. In the special case of Khibina (Russia), carbonatites are associated with peralkaline foid syenites that probably formed under very reduced conditions. There, the carbonatites are of carbothermal rather than magmatic origin and the reduced silicate rocks derive from a different mantle source (Kramm and Kogarko, 1994).

Despite all limitations and exceptions, the contrasting redox conditions between alkaline complexes that contain carbonatites and many alkaline complexes devoid of carbonatites is striking (Fig. 4) and is probably related to highly variable redox conditions in the mantle. Alkaline complexes that show reduced crystallization conditions are generally characterized by the absence of carbonatites, presumably because of the absence of oxidized and carbonate-involving metasomatism processes. Alkaline silicate rocks that are genetically associated with carbonatites, however, belong to the most oxidized alkaline rocks at continental settings, consistent with the prerequisite of an oxidized, carbonate-bearing mantle.

Declaration of competing interest

The authors declare that they have no known competing financial interests or personal relationships that could have appeared to influence the work reported in this paper.

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Appendix A. Supplementary material

Supplementary material related to this article can be found online at <https://doi.org/10.1016/j.epsl.2019.116041>.

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