

Zirconium-bearing accessory minerals in UK Paleogene granites: textural, compositional, and paragenetic relationships

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Abstract. The mineral occurrences, parageneses, textures, and compositions of Zr-bearing accessory minerals in a suite of UK Paleogene granites from Scotland and Northern Ireland are described. Baddeleyite, zirconolite, and zircon, in that sequence, formed in hornblende + biotite granites (type 1) and hedenbergite–fayalite granites (type 2). The peralkaline microgranite (type 3) of Ailsa Craig contains zircon, dalyite, a eudialyte-group mineral, a fibrous phase which is possibly lemoynite, and Zr-bearing aegirine. Hydrothermal zircon is also present in all three granite types and documents the transition from a silicate-melt environment to an incompatible elementrich aqueous-dominated fluid. No textures indicative of inherited zircon were observed. The minerals crystallized in stages from magmatic through late-magmatic to hydrothermal. The zirconolite and eudialyte-group mineral are notably Y+REE-rich (REE signifies rare earth element). The crystallization sequence of the minerals may have been related to the activities of Si and Ca, to melt peralkalinity, and to local disequilibrium.

1 Introduction

Zr-bearing accessory minerals form over a very large P-T (pressure and temperature) range and in a wide variety of lithologies. The stability of these minerals relative to their host rock compositions and to other Zr-bearing phases is complex and variable (e.g., Marks et al., 2011; Andersen et al., 2013). Relatively little is known of the factors controlling the distribution of the phases, and the textural relationships between them are often poorly documented. Critical parameters in determining which phase will form include whole-rock composition and the Si and Ca activities. The Zr silicates are also prone to being altered by hydrothermal processes, commonly being replaced by other Zr silicates (e.g., Wu et al., 2015).

Attempting to understand the controls over mineral stability is particularly valuable in studies of specific igneous complexes, which can be taken as quasi-closed systems. Several studies have traced the textural and compositional variations in Zr-bearing minerals during the evolution of specific silica-oversaturated alkaline complexes, including Birkett et al. (1992), Kynicky et al. (2011), Estrade et al. (2014, 2018), and Sarangua et al. (2019). In this paper, we report on the occurrence of Zr-bearing accessory minerals in some UK Paleogene granites.

In previous studies of UK Paleogene granites, zircon (ZrSiO₄) has been recorded optically in several of the granites, e.g., the Beinn Dearg Mhor Epigranite, the Marsco Epigranite, the Glas Bhein Mhor Epigranite (Bell, 1966), and the Glamaig Epigranite (Thompson, 1969). Zirconolite (CaZrTi₂O₇; referred to as zirkelite) has been reported only in a pegmatitic pod from the island of St. Kilda (Harding et al., 1982; Fowler and Williams, 1986). The type locality of bazirite (BaZrSi₃O₈) is Rockall Island, where it occurs in an aegirine–riebeckite granite (Young et al., 1978). Elpidite (Na₂ZrSi₆O₁₅•3H₂O) and a eudialyte-group mineral (Na₁₅Ca₆(Fe²⁺,Mn²⁺)₃Zr₃(Si,Nb)(Si₂₅O₇₃)(O,OH,H₂O)₃ (Cl,OH)₂) have been recorded from Rockall Island (Young et al., 1978), and eudialyte is also known from the Carlingford Complex, Northern Ireland (Nockolds, 1950).

No attempt has yet been made, however, to examine the compositional ranges of the minerals or to see them as a paragenetic sequence. The specific aims of the present study are (1) to document more fully the occurrence of Zr-bearing minerals in a representative suite of the Paleogene granites; (2) to relate the occurrences and chemical compositions to the compositions of the host rocks, although this goal is limited by the lack of modern geochemical data for many of the granites; (3) to determine the crystallization sequence of the Zr-bearing phases in the granites and to attempt to define the controlling factors; and (4) to report on the occurrence of dalyite (the first record in the UK), a eudialyte-group mineral and possible lemoynite in the Ailsa Craig microgranite.

2 Geological setting of the UK Paleogene granites

During the Paleogene Epoch, the so-called Iceland plume caused thermal uplift over an area some 2000 km across, stretching from northern Norway in the north to Scotland in the south. The resultant igneous rocks constitute the North Atlantic Igneous Superprovince and have a volume of 6- 10×10^6 km³. The British section of the superprovince, the Paleogene Hebridean Igneous Province (Emeleus and Bell, 2005), consists of igneous centers exposed along the west coast of Scotland and in Northern Ireland and as several submerged bodies. The centers consist of lava piles, intrusive central complexes, and sets of linear and radial dike swarms. The central complexes contain a wide range of rock types, including peridotite, olivine gabbro, granites, and minor volumes of rocks with intermediate compositions. This report concerns granitic rocks from six complexes (Fig. 1). The ages of the granites are in the range 58 ± 1 Ma (Emeleus and Bell, 2005). The samples used here are a subset of those used in studies of minerals of the chevkinite group (Macdonald et al., 2013a) and apatite supergroup (Macdonald et al., 2013b) (Table 1).

The granites can be broadly grouped into three types on the basis of their ferromagnesian mineral assemblages (Bell, 1982): type 1, biotite \pm amphibole; type 2, hedenbergitefayalitic olivine; and type 3, alkali amphibole. The allocation of the granites into the three lithological types is given in Table 1. Although the intrusions are invariably referred to as granites in the literature, the usage followed here, they vary petrographically from alkali-feldspar granites to potassic adamellites (e.g., the Glamaig Granite; Table 1) (Thompson, 1983). The granites range from metaluminous to peralkaline, the mineralogical differences being reflected in the geochemical characteristics; the average peralkalinity indices (PIs; mol. $(Na_2O+K_2O)/Al_2O_3$) in types 1, 2, and 3 are 0.88, 0.95, and 1.05, respectively (Bell, 1982). It may be noted that the most peralkaline of the Paleogene granites is Rockall Island, with a PI > 1.7 (Hawkes et al., 1975); the granite almost certainly crystallized from a pantelleritic magma.



Figure 1. Location of Paleogene granitic bodies studied here. Detailed geological maps of the bodies may be found in Emeleus and Bell (2005). The samples used here are a subset of those used in studies of minerals of the chevkinite group (Macdonald et al., 2013a) and apatite supergroup (Macdonald et al., 2013b) (Table 1).

3 Conditions of formation of the granites

There are several difficulties in estimating the P, T, and fO_2 conditions under which the granites crystallized. For example, the feldspars are commonly exsolved, albitized, or hydrothermally altered, and FeTi oxides are exsolved and/or oxidized. Macdonald et al. (2013b) used the apatite saturation temperature models of Harrison and Watson (1984) and Piccoli and Candela (2002) for the metaluminous granites (types 1 and 2) to suggest that they crystallized in the range 772-900 °C. These temperatures are broadly in line with liquidus temperatures determined experimentally by Thompson (1983) for two Skye granites: 910–930 °C for the Glamaig Granite and 752 °C for the Southern Porphyritic Felsite. On the basis of oxygen isotope data, Ferry (1985) estimated that the initial stages of crystallization of the Beinn an Dubhaich Granite (type 1) occurred at 740-800 °C, with latestage crystallization at 660-700 °C. Thompson (1983) suggested that the Skye granites crystallized at $pH_2O < 1$ kb, in keeping with geological evidence that they were shallow intrusions (Thompson, 1969; Emeleus and Bell, 2005).

4 Samples and analytical methods

New whole-rock analyses of five granites were made at Activation Laboratories, Ltd., Ancaster, Ontario, Canada (Table 2). All samples were crushed and powdered with mild steel before analysis. Inductively coupled plasma mass spectrometry (ICP-MS) and inductively coupled plasma optical

					Zirc	on					
Intrusion	Sample number	Lithology and granite type	М	LSM	HT1	HT2	Bdy	Zrc	EGM	Dly	FZr
Skye											
Skye: Eastern Red Hills Centre											
Creag Strollamus Granite	H1226	Porph. bi \pm amp granite (1)	\checkmark		\checkmark	\checkmark					
Beinn na Cro Granite	H1200	$Bi\pm amp$ granite (1)	\checkmark	\checkmark							
Glas Bheinn Mhor Granite	GBME	Porph. bi \pm amp granophyre (1)	\checkmark								
Skye: Western Red Hills Centre											
(Northern) Porphyritic Felsite	NPF(sheared)#1	Porph. microgranite (2)	\checkmark			\checkmark					
Loch Ainort Granite	H1554	Porph. fa-fh granophyre (2)	\checkmark	\checkmark	\checkmark	\checkmark					
Maol na Gainmich Granite	14709	Riebeckite granite (3)	\checkmark			\checkmark					
Glamaig Granite	GLE2	Porph. bi±amp granite (1)	\checkmark	\checkmark		\checkmark		\checkmark			
Skye: Strath na Creitheach Centr	'e										
Meall Dearg Granite	MD	Porph. fa-fh microgranite (2)	\checkmark	\checkmark							
Ruadh Stac Granite	RS1	Riebeckite granophyre (3)			\checkmark						
Rum											
Western Granite	RUM	Bi±amp granophyre (1)	\checkmark		\checkmark		\checkmark				
Ardnamurchan											
Granophyre, Centre 2	14724	Fa-fh granite (2)	\checkmark				\checkmark	\checkmark			
Grigadale Granophyre	13138	Pyroxene granophyre (2)	\checkmark		\checkmark	\checkmark	\checkmark	\checkmark			
Mull											
Beinn a'Ghraig Granophyre	13274	Fa-fh granite (2)	\checkmark								
Toll Doire Granophyre	44858	Fa-fh granophyre (2)	\checkmark	\checkmark							
Ailsa Craig											
Ailsa Craig granite	DV24828	Riebeckite microgranite (3)			√	✓			√	~	✓
Mourne Mountains											
G1 cumulate	G1 cum	Rich in mafics and accessories	\checkmark								

Table 1. Location and lithology of the Paleogene granites used in this study and occurrences of Zr minerals.

✓, present; bi, biotite; amp, amphibole; fa, fayalite; fh, ferrohedenbergite; porph., porphyritic; M, magmatic; LSM, late-stage magmatic; HT1, hydrothermal type 1; HT2, hydrothermal type 2. Bdy, baddeleyite, Zrc, zirconolite; EGM, eudialyte-group mineral; Dly, dalyite; FZr, Zr-bearing fibrous phase. For (1), (2), and (3) see text for explanation. Accessory minerals identified in these samples are given in the Supplement.

emission spectrometry (ICP-OES) were done using a lithium metaborate–tetraborate fusion procedure before sample dissolution or a multi-acid dissolution. Analytical methods employed were ICP-OES for the major elements and Be, Cu, Ni, Zn, Ba, Sr, V, and Zr, instrumental neutron activation analysis for Cr, Sc, and As, and ICP-MS for all other trace elements. Loss of ignition (LOI) was determined at 1000 °C. Estimated relative uncertainties (1σ) were $\leq \pm 5\%$ at 10 times the detection limit and $\pm 20\%$ at the detection limit. Although, many of the analytical techniques used for the Ailsa Craig element analyses (Table 2) are outdated, we include this analysis for discussion of the Zr-bearing phases in the Ailsa Craig granite and also note that no detailed petrographic description is given for the sample (Govindaraju, 1987).

Thin sections were examined by a Hitachi SU5000 Schottky field emission scanning electron microscope (FE-SEM) using back-scattered electron (BSE) and secondary electron (SE) imaging to define points for analysis and for photomicrographs. The FE-SEM was equipped with an EDAX Octane 30 mm² silicon drift detector (SDD). Operating conditions for mineral identification and data collection by energydispersive X-ray spectroscopy (EDS) using the SDD were 15 or 20 keV, 50 spot position (~ 4 nA probe current), and 200 s live counting time. EDS qualitative and semi-quantitative spectra were acquired and used to plan electron microprobe microanalysis (EPMA) procedures. Concentrations were calculated from the EDS count data using the EDAX factory standardization and either the eZAF or PeBaZAF quantification scheme. Cathodoluminescence (CL) imaging was obtained using a Hitachi ultra variable-pressure detector (UVD) operated at zero bias at high vacuum on the Hitachi FE-SEM. CL collection conditions were 10 or 15 keV, 50 spot position (\sim 4 nA probe current), and 18 mm working distance. Quantitative electron microprobe analyses of major and minor elements were obtained with a JEOL JXA-8900 five spectrometer, a fully automated electron microprobe using wavelengthdispersive X-ray spectroscopy (WDS) at Reston, VA. Electron probe microanalysis experiments were made at 15 keV accelerating voltage, 30 nA probe current (Faraday cup), and counting times of 20-120 s, using a 1 µm or defocused probe spot. X-ray lines used for analyses were as follows: $K\alpha =$ Si, Ti, Al, Fe, Mg, Ca, Mn, K, Na, Cl, F, P; $L\alpha = Ba$, Ce, Sr, Er, La, Y, Yb, Nb, Hf, Ta, Zr; $L\beta = Pr$, Nd, Sm, Gd, Dy, Eu; $M\alpha$ = Th; and $M\beta$ = U, Pb. Detection limits (3 σ) (as wt %) were as follows: K, Na = 0.01; Si, Al, Fe, Mg, Ca, Mn = 0.02; Ti = 0.03; Sr, P, Cl = 0.04; Ba = 0.05; Pb, Zr, Th, U, Ta = 0.06; Hf, Nb, Y, F = 0.08; La, Ce, Pr, Nd, Er, Yb = 0.10; Sm, Gd = 0.12; Eu = 0.14; and Dy = 0.15. Standards

Table 2. Chemical compositions of selected granites with Ailsa Craig.

Granite	G. Bheinn Mh	Meall Dearg	Rudha Stac	N. Porphyr.	Glamaig	Ailsa Craig*
Sample no.	GBME	MD	RS1	NPF (sheared)	GLE2	
SiO ₂ (wt %)	72.35	69.28	69.70	74.17	68.57	70.35
TiO ₂	0.41	0.47	0.34	0.28	0.64	0.11
Al_2O_3	12.58	12.57	12.43	11.65	13.60	14.70
Fe ₂ O ₃ *	4.24	5.99	4.61	3.71	5.41	2.53
MnO	0.08	0.15	0.14	0.04	0.12	0.06
MgO	0.24	0.11	0.05	0.13	0.61	0.03
CaO	1.05	1.58	0.38	0.69	2.41	0.34
Na ₂ O	4.14	4.84	4.81	3.40	3.37	6.54
K ₂ O	4.53	4.92	5.53	4.51	4.79	4.49
P_2O_5	0.03	0.02	< 0.01	< 0.01	0.19	0.01
LOI	0.58	0.63	0.62	0.08	0.73	0.37
Total	100.23	100.56	98.61	98.66	100.44	99.53
Peralk Ix	0.93	1.06	1.12	0.90	0.79	1.06
Be (ppm)	3	3	4	1	2	12
Ba	979	1583	892	385	1843	55
Sr	71	107	30	40	132	3
Rb	135	109	118	70	85	152
Cs	0.9	1.4	2.5	0.1	0.7	3
V	10	< 5	< 5	11	16	3
W	1.4	< 0.5	31	< 0.5	3.2	1.5
Co	3	< 1	< 1	2	4	0.2
Cr	< 5	< 5	< 5	< 5	7	3.4
Cu	8	5	3	132	12	4
Ni	2	2	< 1	1	2	1.5
Zn	106	197	208	34	104	224
Pb	15	14	20	6	8	39
Sn	4	3	4	2	2	13
As	1.2	< 0.5	1.1	0.9	< 0.5	2.3
TI	0.66	0.59	0.50	0.36	0.42	0.9
Zr	514	567	671	394	326	780
Ht	11.3	12.4	14.1	9.6	6.7	27.9
Nb	12.6	22.1	22.8	18.5	7.9	110
la	1.30	1.32	1.30	1.06	0.72	6.4
Ga	24	24	25	22	20	39
Ge	1.5	1.6	1.3	1.2	I.I	na
lh U	8.79	6.27	0.80	5.76	5.10	18.5
U Sa	1.91	1.11	1.28	1.05	1.1/	4.0
SC V	4./	4.9	1.0	1.3	9.5	0.11
I La	83.8 50.7	120	108	70.4	45.4	184
La	59.7 121	//.1	98.8	80.4	49.1	59 154
Dr	121	100	202	20.3	90.J 11.6	134
ri Nd	14.7	19.0	24.3	20.3	45.1	02
INU Sm	12.2	16.3	00.0	13.3	43.1	92
SIII	15.5	10.3	17.3	13.7	9.18	24.2
Eu	2.50	5.5	2.29	1.62	2.70	26
Gu Th	13.3	10.2	13.8	11.0	0.J 1 D0	20 1 0
	2.55	5.11	2.90	2.04	1.29	4.8
Dy Uo	14./	19.7	18./	12.8	8.02 1.65	29 6 E
110 Er	5.04	4.23	3.92 11.6	2.03	1.03	0.5
Ef Tm	9.12	12./	11.0	/.8/	4.83	17.7
1 III Vh	1.30	1.80	1./3	1.18	0.091	2.0
10 Lu	9.02	12.2	11.4	0.08 1.20	4.33	1/.4
டய	1.43	1.71	1.//	1.50	0.713	2.43

 $\overline{ \text{Granites: G. Bheinn Mh., Glas Bheinn Mhor; N. Porphyr, northern Porphyritic. Fe}_{2O_3}* all Fe as Fe}^{3+}. Peralk Ix = peralkalinity index (mol. (Na}_2O+K_2O)/Al}_{2O_3})). * Composition from Govindaraju (1987); na = not analyzed.$

were appropriate synthetic or natural materials available in the Reston Microbeam Laboratory (Huebner and Woodruff, 1985); Zr, Hf, and Th standards were provided through the courtesy of John M. Hanchar (Memorial University of Newfoundland). The analyses were corrected for electron beammatrix effects and instrumental drift and dead time using a Phi-Rho-Z scheme (CITZAF; Armstrong, 1995) as supplied with the JEOL JXA-8900R electron microprobe. The relative accuracy of the analyses, based upon comparison between measured and published compositions of standard reference materials available in the Reston Microbeam Laboratory, is $\sim 2\%$ for element concentration > 1 wt % and $\sim 5\%$ -10 % for element concentrations < 1 wt %. The complete data sets for zircon and other Zr-bearing phases are given in the Supplement.

5 Results – modes of occurrence and compositions of zircon

Zircon is a ubiquitous accessory mineral in granitoids except in strongly peralkaline and agpaitic compositions in which other Zr species may predominate. Its stability and lack of common lead has made it a valuable tool in geochronology, especially with the advent of high-spatial- and high-massresolution ion probes (e.g., SHRIMP) and laser ablation ICP-MS techniques (e.g., Harley and Kelly, 2007). Zr, Hf, and O isotope studies are used also to trace magma genesis, metamorphic reactions, and alteration mechanisms (e.g., Liu et al., 2009; Valley et al., 2010; Ibañez-Mejia and Tissot, 2019).

Zircon ($ZrSiO_4$), an orthosilicate, conforms to the formula ABO₄, in which A represents the relatively large Zr ion in eight-fold coordination with O, and B represents the Si ion in tetrahedral coordination with O. Analyses of zircons in this report have been recalculated on the basis of the unit cell content of 16 oxygens (Deer et al., 1982; Finch and Hanchar, 2003). Zircon can have limited element substitution except for Hf in which complete solid solution exists between zircon and hafnon (HfSiO₄) (Ramakrishnan et al., 1969; Speer and Cooper, 1982). Hoskin and Schaltegger (2003) proposed the following trace-element distributions in the structure of zircon: four-fold site – Si^{4+} , P^{5+} ; eight-fold site – Zr^{4+} , Hf^{4+} , Y^{3+} , REE³⁺ (rare earth element), U⁴⁺, Th⁴⁺, Pb⁴⁺(only radiogenic), Ti⁴⁺, Sn⁴⁺, Sc³⁺, Al³⁺, Nb⁵⁺, Ta⁵⁺; and interstitial site - Li⁺, Mg²⁺, Fe²⁺, Mn²⁺, Al³⁺, Fe³⁺. Element substitution can be a simple isovalent replacement of Zr by other tetravalent cations or coupled if divalent and trivalent cations are incorporated (Hoskin and Schaltegger, 2003).

Hoskin and Schaltegger (2003) summarized the following simple and coupled element substitutions commonly observed in zircon: (1) simple substitutions: $Hf^{4+} = Zr^{4+}$, $(U^{4+}, Th^{4+}, Ti^{4+}, Sn^{4+}) = Zr^{4+}$, $(OH)_4^{4-} = SiO_4^{4-}$; (2) coupled substitutions: a – at one structural site: $(Y,REE)^{3+} +$ $(Nb,Ta)^{5+} = 2Zr^{4+}$; b – at two structural sites: $(Y,REE)^{3+} +$ $P^{5+} = Zr^{4+} + Si^{4+}$, $(Sc)^{3+} + P^{5+} = Zr^{4+} + Si^{4+}, M^{n+} +$ $n(OH)^-$ + (4–n)H₂O = Zr⁴⁺+ (SiO4)⁴⁻ (where M = metal cation and n = integer); and c – at an interstitial site: (Mg,Fe)²⁺(int) + 3(Y,REE)³⁺+ P⁵⁺ = 3Zr⁴⁺+ Si⁴⁺, (Al,Fe)³⁺(int) + 4(Y,REE)³⁺+ P⁵⁺ = 4Zr⁴⁺ + Si⁴⁺. The most common substitutions in zircon are Hf⁴⁺ = Zr⁴⁺ and the "xenotime" substitution (Y,REE)³⁺ + P⁵⁺ = Zr⁴⁺ + Si⁴⁺ (Speer, 1982).

Although, xenotime substitution appears to be the dominant trace-element substitution mechanism in many zircons, Hoskin and Schaltegger (2003) point out that the common occurrence of REEs (mol %) in excess of P (mol %) in natural zircon indicates that other mechanisms also operate. In the studied zircons, many have REEs (mol %) in excess of P (mol %); thus, other substitution mechanisms in addition to the xenotime type were operative. We have used the sum of Σ REE+Y+P apfu (atom per formula unit; not including U or Th) as a measure of the degree of trace-element substitution and deviation from ideal zircon.

In the studied samples, zircon is the main Zr mineral except in the Ailsa Craig peralkaline granite in which dalyite and a eudialyte-group mineral predominate. On the basis of texture primarily and composition, the zircon population is classified into four groups: magmatic, late-stage magmatic, hydrothermal type 1, and hydrothermal type 2.

5.1 Magmatic zircons

Magmatic zircons are defined as those that crystallized from a silicate melt during early zircon saturation. Magmatic zircons have been identified in all samples except the Ruadh Stac Granite and Ailsa Craig microgranite (Table 1). Harding (1983) reports no zircons found in his eight Ailsa Craig samples, whereas Harrison et al. (1987) report zircons in the crush but do not give a detailed petrographic description. All the magmatic zircons are subhedral to euhedral prisms (Fig. 2a, b, d, g), with common combinations of prism ({100} and $\{110\}$) and pyramid forms ($\{211\}, \{101\}$ and $\{301\}$). Their length-to-width ratio is usually 2:1 to 4:1, although a few were more acicular and have ratios of 6:1 to 8:1 (Fig. 2h). They range in size from 10–500 µm. All have some degree of oscillatory and/or sector zoning (Fig. 2c, e, f, g, i), typical of igneous magmatic zircon (Hoskin and Schaltegger, 2003). Silicate melt (Fig. 2a, b, g) and apatite (Fig. 2d, h) are the most common inclusions, although britholite (Fig. 2b), quartz, pyrrhotite, chevkinite-(Ce), titanomagnetite, and Kfeldspar inclusions are also present. We did not have sufficient sample to attempt any microthermometric studies of the silicate-melt inclusions but tentatively identified them by their round or ovoid shape (Fig. 2b, g); a few had a negative crystal shape (Fig. 2a). All had low BSE response, as would be expected for trapped high SiO₂ compositions. The magmatic zircon compositions do not differ appreciably from ideal ZrSiO₄ (Tables 3, 4). HfO₂ ranges from 0.70 wt %-1.37 wt % and Zr/Hf w/w from 42–81, although 95 % fall between 42-65 (Fig. 3a). Trace-element substitution is minor, and $\Sigma REE+Y+P$ varies from 0.018–0.146 apfu (16 oxygen basis) (Fig. 4a). The contents of ΣREE , Y_2O_3 , ThO₂, and UO₂ are generally low (Figs. 5a and 6a). There is no significant compositional difference among all the magmatic zircons. Additionally, we observed no textures characteristic of inherited zircons in any of the samples (e.g., Paterson et al., 1992; Aleinikoff et al., 2006). This lack of an inherited zircon component points to an origin of the granites being dominantly by fractional crystallization of mafic magmas with insignificant crustal contamination, as suggested by Gamble et al. (1992, 1999) for Paleogene granites of Slieve Gullion and Mourne Mountains.

5.2 Late-stage magmatic zircons

In a few cases, anhedral magmatic zircons are seen to have grown in the interstices and spaces between crystals formed earlier. These relatively late-crystallizing zircons have oscillatory and/or sector zoning, contain silicate-melt inclusions (Fig. 7a, b, c), and are compositionally indistinguishable from earlier magmatic zircons in the same sample. HfO₂ varies from 0.99 wt %–1.33 wt %, and Zr/Hf w/w ranges from 43–58 (Fig. 3b; Table 5). Trace-element substitution is minor (0.031–0.059 apfu; Fig. 4b), and the abundances of Σ REE, Y₂O₃, ThO₂, and UO₂ are low (Figs. 5b and 6b; Table 5).

5.3 Hydrothermal zircons type 1 and type 2

For this study, hydrothermal zircons are defined as those precipitated from an aqueous-dominated, solute-rich fluid of magmatic origin. Such fluids typically exsolve from granitoid bodies during the later stages of cooling. Two hydrothermal zircon types are distinguished primarily on texture. We did not observe any textures that suggested that the hydrothermal zircons resulted from the alteration of magmatic zircon. Hydrothermal type 1 zircons are either anhedral, filling interstices and spaces between earlier formed crystals, or are uncommon euhedral crystals in vugs, essentially devoid of inclusions and enriched in incompatible elements such as Y, Th, U, and Σ REE. Hydrothermal type 2 zircons are filled with fluid and solid inclusions and are, in many cases, less enriched in incompatible elements than type 1, except those from Ailsa Craig. Both types, except for Ailsa Craig type 2, have EPMA sums that suggest no OH⁻ or H₂O component.

5.3.1 Hydrothermal type 1 zircons

Hydrothermal type 1 zircons were identified in six samples (Table 1). The category includes zircon that has replaced baddeleyite (see below). The zircons are anhedral, filling interstices and cracks (Fig. 8a, c), and do not display oscillatory or sector zoning but appear patchy or diffusely zoned in CL or BSE images (Fig. 8b, c). In a few cases euhedral crystals occur in vugs (Fig. 8d). Type 1 has a large range of HfO₂ (0.58 wt %–2.79 wt %), and Zr/Hf w/w varies from 18–91, but most are between 37–70 (Fig. 3c; Tables 6, 7). Traceelement substitution is significantly greater than the magmatic type, and $\Sigma REE+Y+P$ varies from 0.149–0.695 apfu (Fig. 4c; Tables 6 and 7). The simple substitution, U⁴⁺+ Th⁴⁺ = Zr⁴⁺, is also present. There is a positive correlation between Y₂O₃ and ΣREE (Fig. 5c) and ThO₂ and UO₂ (Fig. 6c), with ThO₂ usually greater than UO₂. In one case, the textures suggest that the fluids forming a hydrothermal type 1 zircon have partially altered an earlier chevkinite-(Ce) crystal (Fig. 8a) but do not appear to have altered the other adjacent phases.

5.3.2 Hydrothermal type 2 zircons

Hydrothermal type 2 zircons are fluid- and solid-inclusionrich. Those in Ailsa Craig are placed in type 2 because they have cavities but no fluid inclusions sensu stricto (Roedder, 1984). Type 2 are most abundant in the Skye Western Centre and Eastern Red Hills Centre and less common in the Ardnamurchan Grigadale Granophyre (Table 1). Figure 9a illustrates the transition, sometimes observed, between type 1 and type 2. A type 1 zircon has begun to fill a space between magnetite grains, when the zircon began to trap both fluid and solid inclusions (Fig. 9a). In other cases, type 2 zircon precipitated around an earlier-formed magmatic zircon, yet it displays a euhedral form (Fig. 9b, c). Groups of inclusion-rich zircons also occur at grain boundaries (Fig. 9e) and fill interstices among crystals (Fig. 9f). We identified (EDS) xenotime, thorite, fergusonite, calcite, REE+Y fluorcarbonate, U,Y,Si carbonate, britholite, Fe oxide, quartz, chlorite, probable hingganite-(Y), and probable hellandite-(Y) as solid inclusions (e.g., Fig. 9d1, d2 ,d3). There are no inclusion assemblages to suggest that the included phases were daughter minerals (Roedder, 1984), and so they are interpreted to be accidentally trapped solids. Figure 9a shows the transition between hydrothermal type 1 with a high incompatible element content (pt. $1 - Y_2O_3$ 5.04, ThO₂ 1.03 wt %; Table 6, analysis H1226_Z1core) and hydrothermal type 2 zircon with a low incompatible element content (pt. $2 - Y_2O_3$) 0.37, ThO₂ 0.19 wt %; Table 8, analysis H1226_Z1rim). The presence of Y- and Th-bearing trapped minerals in type 2 indicates that the zircon-forming fluid was precipitating Y- and Th-rich phases, hence depleting the fluid and zircon in those elements.

The Ailsa Craig hydrothermal type 2 zircons have a distinct habit and composition different from the other type 2 zircons. The Ailsa Craig zircon fills vugs and has cavities lined with a botryoidal texture (Fig. 9g). HfO₂ varies between 0.91 wt %–1.26 wt % and Zr/Hf w/w is in a narrow range between 32–43 (Fig. 3d; Table 9). Trace-element substitution ($\Sigma REE+Y+P$) varies from 0.38–0.91 apfu and has a scattered distribution (Fig. 4d). The zircons have high Y₂O₃ and ΣREE contents (Fig. 5d), but in contrast to all other studied zircons, they have relatively high LREE (La–Gd REE) contents and compared to other hydrothermal type 1 and 2

Skye.
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Table 3.

ntrusion Analysis V =																
Malysis V =	Creag Strolls	amus Granite	Beinn na	Cro Granite	Glas Bheinn	Mhor Grn.	N. Porphyri	tic Felsite	Loch Aino	rt Granite	Maol na Ga	inmich Grn.	Glamai	g Granite	Meall De	carg Grn.
	H1226_Z2 4	H1226_Z4 3	H1200_Z3 3	H1200_Z2b 3	GBME_Z5 6	GBME_Z4	NPFx1_Z1c 3	NPFx1_Z3	H1554_Z1c 1	H1554_Z2 3	14709_Z2 3	14709_Z5 3	GLE2_Z1 3	GLE2_Z7a 2	MD_Z5	MD_Z3
iO2 (wt %)	32.62	32.40	32.49	32.40	32.63	32.58	32.50	32.53	32.53	32.31	32.68	32.38	32.86	32.48	32.51	32.42
102	lbd	lbd	0.04	lbd	lbd	lbd	lbd	lbd	0.03	lbd	lbd	lbd	0.03	lbd	lbd	pq
eO*	0.03	0.19	0.29	0.12	0.16	0.20	0.26	0.14	0.36	0.02	0.35	0.04	0.25	0.21	0.12	0.11
AgO	Ipq .	Ipq .	lbd	Ipq .	pdl	lpq .	lbd	Ipq .	lbd 	lbd	Ipq .	lbd 0.00	pdl	0.02	Ipq	pq ;
a0	lbd	lbd	lbd	lpd	0.02	lpq	0.02	pql	lbd	lbd 1	Ibd	0.03	0.02	0.03	0.03	pd :
102	65.51	63.63	64.36	64.14	65.98	66.39	62.79	65.68	64.61	63.08	65.17	64.47 2.22	65.95 22	64.48	65.11	64.7
IfO2	1.24	1.22	1.30	1.31	1.14	1.20	1.10	0.91	1.03	1.01	0.97	0.83	1.25	1.15	0.92	0.9(
h0 ₂	0.09	0.08	lþd	lbd	lbd	lpq	lbd	0.06	lbd	0.10	lpq	0.06	lþd	0.13	lbd	þq
JO ₂	lbd	lbd	0.16	0.07	lbd	0.05	lbd	lbd	lbd	lbd	0.07	lbd	lbd	lbd	lþq	pq
Vb2O5	lbd	lbd	lbd	lbd	lbd	lpq	lbd	lbd	lbd	lbd	lbd	lbd	0.09	lbd	lbdl	pq
205	0.12	0.16	0.18	0.22	0.12	0.14	0.16	0.18	0.16	0.23	0.11	0.22	0.11	0.30	0.14	0.22
(203	0.29	0.65	0.85	0.75	0.21	0.15	0.57	0.72	0.73	1.43	0.63	0.92	0.10	0.66	0.53	0.68
e.03	lpd	lpd	lbd	0.11	hdl	lpd	0.10	lbd	hdl	lbd	hdl	lbd	lbd	0.17	hdl	pq
	lpd	0.14	hdl	hdl	lpd	- IP4	Pdl	lpd	lpd	0.10	0 11	lрч	РЧ	010	hdl	4
140°	Ipq	hdl	0 11	lpq	Pdl	- Ipq	Ipq	Ipq	0.12	hdl	hdl	Ipq	Ipq	PHI PHI	IP4	2 P4
677	100	- IP-1	11.0	100	IP-1	IP4	191	0.12	11-1	IP4	100	IP4	100		510	33
11203	IDO I		IDO 1	IDO I	IDO I	IDO 1	100	01.0	IDO 1	ID O	IDO I	ID O	ID O		71.0	33
1d2 U3	IDQ .	0.12	ID .		ID GI	IDd .	CI.U 	0.19	IDdi .	IDd .			Dd1	0.12	ID q	ра : Ба
0y203	Ipq	lbd.	lbd	0.22	Ipq .	Ipq .	lbd.	Ipq .	lbd.	lbd	Ipq	0.15	0.18	lbd	Ipq .	pq .
1203	pql	pql	0.15	0.20	pq	lpq	pdl	pql	pdl	0.25	pq	lbd	lbd	0.18	pq	pq
(b203	lbd	0.10	0.21	0.15	lbd	lpq	lbd	lbd	0.18	0.22	0.12	0.15	lbd	0.20	0.13	0.1
otal	06.90	98.71	100.13	79.69	100.26	100.72	100.64	100.55	99.74	98.75	100.21	99.24	100.83	100.22	09.66	99.2
T/Hf w/w	46.2	45.5	43.1	42.7	50.7	48.2	52.3	62.7	55.0	54.3	58.9	67.8	46.1	48.8	61.9	63.
ormulae on ti	he basis of 16	oxygens														
		2														
	4.001	4.023	3.992	3.998	3.988	3.971	3.969 _	3.975	4.002	4.016	4.001	3.997	3.995	3.985	3.998	4.00
-2+	0.003	0000	0.030	0.012	0.016	0000	0.076	0.015	0.037	0000	0.036	0.004	0.025	0.000	0.012	0.01
ره ۱۹			1	71000			1	10.0		10000	-	100.0	10.0	0.003	10.00	
,	1	I	I	I	0.003	1	0.002	I	I	I	I	0.003	0.002	0.004	0.004	
	3,919	3,853	3.856	3.859	3.932	3.946	3.918	3.913	3.876	3,824	3,892	3,882	3.911	3,859	3,905	3.89
If	0.043	0.043	0.046	0.046	0.040	0.042	0.038	0.032	0.036	0.036	0.034	0.029	0.043	0.040	0.032	0.03
4	0.003	0.002	I	I	I		I	0.002	I	0.003	I	0.002	I	0.004	I	'
ſ	1	I	0.004	0.002	I	0.001	I	I	I	I	0.002	I	I	I	I	
- -	1	I	I	I	I		I	I	I	I	I	I	0.005	I	I	,
_	0.012	0.017	0.018	0.023	0.013	0.015	0.017	0.018	0.016	0.024	0.011	0.023	0.011	0.031	0.015	0.023
	0.019	0.043	0.055	0.050	0.014	0.010	0.037	0.047	0.048	0.095	0.041	0.060	0.007	0.043	0.034	0.045
2	1	I	I	0.005	I		0.005	I	Ι	I	1	I	I	0.007	I	I
-	1	0.006	I	I	I		Ι	I	Ι	0.005	0.005	I	I	0.004	I	I
- PA	1	I	0.005	I	I	1	I	I	0.005	I	I	I	I	I	I	I
, m	1	I	I	I	I	-	I	0.006	I	I	I	I	I	I	0.005	I
þr	1	0.005	I	I	I	1	0.006	0.008	I	I	I	I	I	0.005	I	1
م م	1	I	I	0.009	I	1	I	I	I	I	I	0.006	0.007	I	I	1
	I	I	0.006	0.008	I	1	I	I	I	0.010	I	I	I	0.007	I	1
e	1	0.004	0.008	0.006	I	1		L	0.007	0.008	0.005	0.006	L	0.007	0.005	0.00
Scations	8.00	8.02	8.02	8.02	8.00	8.01	8.02	8.01	8.03	8.02	8.03	8.01	8.01	8.02	8.01	8.01

rusion Western alysis RUM_Z20b	n Granite	Granophyre,	, Centre 2	Grigadale G	anophyre	Beinn a	Ghraig Granoph	lyre	Toll Doire C	iranophyre	G1 cur	nulate	
alysis RUM_Z20b													
= 2	RUM_Z5a 1	14724_Z1 3	14724_Z3 1	13138_Z1a 2	13138_ZB1 2	13274_Z2rim 2	13274_Z2core 2	13274_Z4 3	44858_Z2 3	44858_Z4 3	Glcum_Zla 3	G1cum_Z1b 1	
$\frac{32.67}{2}$ (wt %) 32.67	32.70 0.04	32.24 0.04	32.33 hdl	32.52 hdl	32.24 0.03	32.33 hdl	32.49 hdl	32.42 hdl	32.61 hdl	32.66 hdl	32.64 0.07	32.53 0.09	32.
203 bdl	bdl	bdl	0.07	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	
08.0 *C	0.83	0.31	bdl	0.03	0.82	0.66	0.11	0.08	0.17	0.11	0.49	0.90	
j00	- bdl	bdl	bdl	bdl	bdl	bdl	bdl	- bdl	, bdl	bdl	lpq	, oc	
ບ ບ.ບວ ງ, 64.31	63.24	0.02 65.11	64.17	64.52	0.04 63.67	65.91	65.63	66.13	0.03 65.15	65.18	64.98	0.03 64.57	62
0 ₂ 1.18	0.77	0.87	1.07	1.17	1.33	1.37	1.05	1.10	0.90	1.05	1.10	1.31	;
02 0.06	0.29	- bdl	bdl	0.11	0.25	bdl	0.09	bdl	bdl	bdl	bdl	bdl	
)2 bdl	bdl	Ipq	bdl	0.26	0.11	0.08	0.13	0.10	0.06	0.06	0.09	0.16	
י2U ₅ bdi סבר 0.11	0.36	0.15	0.18	0.14	0.27	0.11	0.17	0.10	0.19	0.15	0.09	0.16	_
$O_3 = 0.34$	0.87	0.38	0.92	0.63	0.57	0.34	0.70	0.31	0.46	0.43	0.10	0.19	_
2O3 bdl	bdl	bdl	bdl	bdl	bdl	0.17	bdl	bdl	bdl	0.11	bdl	bdl	
2O3 bdl	0.10	bdl	bdl	bdl	0.16	bdl	bdl	bdl	bdl	0.15	bdl	bdl	
202 bdl	0.11	bdl	0.10	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	
$b_0 0_3$ bdl bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.27	0.12	bdl	bdl	bdl	bdl	
203 bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.19	bdl	bdl	bdl	bdl	bdl	
¹² O ₃ bdl	0.42	0.16	0.20	bdl	bdl	bdl	0.17	bdl	bdl	bdl	0.18	bdl	
2O3 bdl	0.30	bdl	0.11	0.11	0.11	bdl	0.18	bdl	bdl 0.13	bdl	bdl	bdl	
al 99.49	100.39	99.27	99.61	99.78	99.69	101.12	101.26	100.35	99.71	99.89	99.73	99.92	98.70
/Hf w/w 47.6	72.1	65.4	52.4	48.0	41.9	42.2	54.5	52.7	63.1	54.2	51.4	43.0	4
mulae on the basis of 16	oxygens												
4.020	4.008 0.004	3.981 0.004	3.990	4.008	3.981 0.003	3.942	3.963	3.970	4.002	4.006	4.011	3.995 0.008	3.94
	I	I	0.010	I	I	I	I	I	I	I	I	I	
2+ 0.082	0.085	0.032	1	0.003	0.085	0.067	0.012	0.008	0.018	0.011	0.050	0.092	
	Ι	2021	1	Ι	2000	I	I	1	2021	I	I	2 2 2 2	
0.0U4	00L 6 -	0.003	- د70 د	1 1 0 0	200.0 200.0	1 000	-	3 050	0.004 2 onn	000 5		0.003	۔ د ۱ ۵
0.041	3./ou 0.027	3.921 0.031	0.038	0.041	3.834 0.047	3.920 0.048	3.904 0.037	0.038	3.899 0.032	0.037	0.039	0.046	0.02
0.002	0.008	I	I	0.003	0.007	I	0.003	I	I	Ι	I	I	1
1	I	I	I	0.007	0.003	0.002	0.004	0.003	0.002	0.002	0.003	0.004	
	85U U -	- 0 016	- 0 010	- 0 015		0.006	-	0 010			600 U -	- 0 016	00.
0.022	0.057	0.025	0.060	0.041	0.037	0.022	0.045	0.020	0.030	0.028	0.006	0.012	0.00
	I	I	I	I	I	0.008	I	I	I	0.005	I	I	
1	0.004	I		I	0.007	I	I	I	I	0.007	I	I	
	0.005		C00.0										
		1	-	1	1	1	0.011	0.005	1	1	1	1	1
1	I	I	I	I	I	I	0.008	I	I	I	I	I	
1	0.017	0.006	0.008	I	I		1	I		I			1
1	0.011	I	0.004	0.004	0.004	I	0.007	1	I		0.007	1	
	0 000	1	800.0			1 1	0.007	I		I	0.00 <i>1</i> -		

 Table 5. Late-stage magmatic zircon compositions.

			Skye			
	East. Red Hills	Western	Red Hills Cer	ntre	Strath Creith.	Mull
Intrusion	Beinn Cro	N. Porph. Fels.	Glamaig	Granite	Meall Dearg	Toll Doire
Analysis	H1200 Z1	H1554 Z1a	GLE2 Z8	GLE2 Z9	MD Z1	44858 Z3
N =	4	2	3	2	- 3	4
SiO ₂ (wt %)	32.54	32.87	32.72	32.80	32.78	32.56
TiO ₂	0.13	0.03	bdl	bdl	bdl	bdl
Al_2O_3	bdl	bdl	bdl	bdl	0.03	bdl
FeO*	0.73	0.37	0.10	0.32	0.68	0.06
CaO	0.02	0.02	bdl	0.02	0.06	0.02
ZrO ₂	65.18	66.02	65.18	64.60	65.23	65.04
HfO ₂	1.33	1.12	1.28	1.08	0.99	1.20
ThO ₂	bdl	bdl	bdl	bdl	bdl	0.08
UO_2	bdl	bdl	bdl	0.12	bdl	0.08
P_2O_5	0.10	0.08	0.17	0.17	0.11	0.14
Y_2O_3	0.49	0.10	0.36	0.32	0.49	0.38
La_2O_3	bdl	0.16	bdl	bdl	bdl	bdl
Ce_2O_3	bdl	0.11	bdl	bdl	bdl	bdl
Pr_2O_3	bdl	bdl	bdl	0.14	bdl	bdl
Sm_2O_3	0.17	bdl	bdl	bdl	0.15	bdl
Dy_2O_3	bdl	bdl	bdl	bdl	bdl	0.15
Er_2O_3	0.14	0.10	bdl	bdl	0.14	bdl
Yb ₂ O ₃	0.10	bdl	0.10	0.11	bdl	bdl
Total	100.93	100.99	99.90	99.69	100.66	99.70
Zr/Hf w/w	42.7	51.3	44.5	52.0	57.7	47.5
Formulae on t	the basis of 16 oxy	gens				
Si	3.972	3.995	4.010	4.025	3.996	4.005
Ti	0.012	0.003	-	_		-
Al	0.001			_	0.004	-
Fe ²⁺	0.075	0.038	0.010	0.032	0.069	0.006
Ca	0.002	0.003	-	0.003	0.008	0.003
Zr	3.880	3.913	3.895	3.865	3.878	3.901
Hf	0.046	0.039	0.045	0.038	0.034	0.042
Th				-	-	0.002
U				0.003	-	0.002
Р	0.010	0.008	0.018	0.018	0.011	0.014
Y	0.032	0.006	0.023	0.021	0.032	0.025
La		0.007		-	-	-
Ce		0.005		-	-	-
Pr			-	0.006		
Sm	0.007		-	_	0.006	
Dy				-		0.006
Er	0.005	0.004		-	0.005	
Yb	0.004		0.004	0.004		
Σcations	8.05	8.02	8.00	8.02	8.04	8.01

 $FeO^* = total iron as Fe^{2+}; bdl = below detection limit; N = number of analyses averaged. Nb₂O₅, Nd₂O₃, Eu₂O₃, Gd₂O₃, SrO, PbO, MgO = bdl.$



Figure 2. Magmatic zircons, BSE images except where noted. (a) A 20 μ m euhedral zircon (Zrn) enclosed by chlorite (Chl) with a 2.5 μ m silicate-melt inclusion (MI). The inclusion is enlarged in the upper right; the lower arrow indicates a euhedral-zoned apatite, and the upper arrow points to a TiO₂ phase. Rum, Western Granite, sample RUM. (b) Two ~90 μ m euhedral zircons (Zrn) with silicate-melt (MI), and britholite (Bri) inclusions, enclosed by quartz (Qz) with magnetite (Mag). Mull, Toll Doire Granophyre, sample 44858. (c) CL image of (b) showing sharply defined magmatic oscillatory zoning. (d) An 80 μ m euhedral zircon (Zrn) with apatite (Ap) inclusions enclosed by riebeckite amphibole (Amp) somewhat disaggregated from sample preparation. Skye, Western Red Hills Centre, Maol na Gainmich Granite, sample 14709. (e) CL image of (d) showing fine-scale oscillatory zoning; the left border shows sector zoning. (f) CL image of a 130 μ m euhedral zircon enclosed by K-feldspar and albite (Kfs/Ab) showing fine-scale oscillatory zoning with well-developed sector zoning. Skye, Eastern Red Hills Centre, Creag Strollamus Granite, sample H1226. (g) A CL image of an 80 μ m zircon (Zrn) surrounded by quartz (Qz) and K-feldspar and albite (Kfs/Ab) with well-developed sector zoning. The crystal contains a silicate-melt inclusion (MI) in the upper left, but the similarly shaped dark spot (black arrow) is a non-CL part of the crystal. Skye, Eastern Red Hills Centre, Glas Bheinn Mhor Granite, sample GBME. (h) An acicular 300 μ m (a portion on the left not shown) euhedral zircon (Zrn) on magnetite and intergrown (left portion) with quartz (Qz). Apatite inclusions (Ap) in both the zircon and magnetite. Skye, Eastern Red Hills Centre, Glas Bheinn Mhor Granite, sample GBME. (h) An acicular 300 μ m (a portion on the left not shown) euhedral zircon (Zrn) on magnetite and intergrown (left portion) with quartz (Qz). Apatite inclusions (Ap) in both the zircon and magnetite. Skye, Eastern Red Hills Centre, Glas Bheinn Mhor Gra

zircons have high CaO and Al₂O₃ values (Table 9). The contents of ThO₂ and UO₂ are all < 0.5 wt % except for one value (Fig. 6d). Figure 9g shows that the zircon formed after the precipitation of a pyrochlore-group mineral and the Ailsa Craig hydrothermal type 2 zircons have high Nb₂O₅ contents in contrast to magmatic zircons (Table 9). The EPMA totals varied from 92 wt %–96 wt %. These are hydrous zircons, but the hydrous value (H₂O, OH^{1–}) is not the difference from 100 % as we did not measure all REEs or B, Be, or Li, for example.



Figure 3. Variation in Zr/Hf w/w versus HfO_2 in the four types of studied zircons: (a) magmatic, (b) late-stage magmatic, (c) hydrothermal type 1, and (d) hydrothermal type 2. The legend shown in (a) applies to all four types where appropriate and is valid for Figs. 4, 5, and 6.



Figure 4. Deviation from ideal ZrSiO₄ by trace-element ($\Sigma REE+Y+P$) substitutions in the studied zircons: (a) magmatic, (b) late-stage magmatic, (c) hydrothermal type 1, and (d) hydrothermal type 2.

			Sk	/e			
	Eastern Red	Hills Centre	Western R	ed Hills C.	Strath	na Creith.	Centre
Intrusion	Creag Strolla	amus Granite	Loch Ain	ort Granite	Rua	dh Stac Gra	nite
Analysis	H1226_Z1core	H1226_Z5core	H1554_Z4	H1554_Z5	RS1_Z2	RS1_Z4	RS1_Z5
N =	2	2	3	3	6	4	3
SiO_2 (wt %)	31.57	31.65	31.95	31.68	31.45	31.21	31.48
TiO ₂	0.07	bdl	bdl	bdl	bdl	bdl	bdl
FeO [*]	0.85	0.35	0.52	0.14	0.05	0.09	0.06
CaO	0.11	0.16	bdl	0.02	bdl	bdl	bdl
ZrO ₂	55.65	56.42	59.08	58.36	55.67	55.19	54.92
HfO ₂	1.05	1.10	2.79	1.39	1.27	0.72	0.80
ThO ₂	1.03	1.20	0.94	0.64	0.19	0.25	0.33
UO ₂	0.53	0.53	0.23	0.28	0.06	0.12	0.13
Nb_2O_5	bdl	bdl	0.27	0.10	bdl	bdl	bdl
P_2O_5	0.25	0.18	0.32	0.27	0.73	0.81	0.62
Y_2O_3	5.04	4.53	2.80	3.78	6.02	6.93	7.40
La_2O_3	bdl	0.13	bdl	bdl	bdl	bdl	bdl
Ce_2O_3	0.14	0.10	bdl	bdl	bdl	bdl	0.13
Pr_2O_3	bdl	bdl	bdl	bdl	bdl	0.08	bdl
Nd_2O_3	bdl	bdl	bdl	bdl	bdl	bdl	0.14
Sm_2O_3	bdl	0.15	bdl	bdl	bdl	bdl	bdl
Eu_2O_3	bdl	0.15	bdl	bdl	bdl	0.22	0.20
Gd_2O_3	0.26	0.13	0.18	0.21	0.53	0.73	0.65
Dy_2O_3	0.33	0.19	0.22	0.38	0.81	0.99	1.17
Er_2O_3	0.71	0.66	0.29	0.42	0.62	0.83	0.51
Yb ₂ O ₃	1.18	1.27	0.51	0.96	0.37	0.44	0.32
Total	98.95	98.97	100.34	98.81	98.17	98.64	99.06
Zr/Hf w/w	46.3	44.9	18.5	36.7	38.3	67.1	60.0
Formulae on t	he basis of 16 oxy	gens					
Si	4.026	4.033	3.996	4.013	4.006	3.971	3.996
Ti	0.007	-	_	_	_	_	_
Fe ²⁺	0.091	0.037	0.055	0.015	0.006	0.009	0.007
Ca	0.016	0.022	-	0.003	_	_	_
Zr	3.461	3.507	3.603	3.605	3.459	3.424	3.401
Hf	0.038	0.040	0.100	0.050	0.046	0.026	0.029
Th	0.030	0.035	0.027	0.018	0.006	0.007	0.010
U	0.015	0.015	0.006	0.008	0.002	0.003	0.004
Nb	-	-	0.015	0.006	-	_	-
Р	0.027	0.019	0.034	0.029	0.078	0.087	0.067
Y	0.341	0.307	0.186	0.255	0.408	0.468	0.500
La	-	0.006	-	_	-	_	-
Ce	0.007	0.005	-	_	-	_	0.006
Pr	-	-		-	-	0.004	-
Nd	-	-		-	-	-	0.006
Sm	-	0.007		_	-	-	-
Eu		0.006		_	-	0.010	0.009
Gd	0.011	0.005	0.007	0.009	0.022	0.031	0.027
Dy	0.013	0.008	0.009	0.016	0.033	0.041	0.048
Er	0.028	0.027	0.011	0.017	0.025	0.033	0.020
Yb	0.046	0.049	0.019	0.037	0.014	0.017	0.012
Σ cations	8.16	8.13	8.07	8.08	8.10	8.13	8.14

Table 6. Representative hydrothermal type 1 zircon compositions from Skye

 $eO^* = total iron as Fe^{2+}$; bdl = below detection limit; N = number of analyses averaged. PbO, SrO, MgO, Al₂O₃ = bdl.

compositions.
zircon

type
hydrothermal
. Representative
5
Table (

$\label{eq:relation} \begin{tabular}{ c c c c c } \hline Intrusion \\ Analysis \\ N = \\ N $	29 RUM_Z13 29 RUM_Z13 54 32.19 6d bdd 18 0.23 6d bdd 18 0.23	Western Granite RUM_Z3b 1	e RUM_Z12	RUM_Z21	W ₆ RUM_ZB10	stern Granite RUM_ZB2	(replacing badd RUM_ZB8	leleyite = ZB)	TD1 ZD11	Grig: 13138_Z7	adale Granopl 13138_Z3	13138_Z8	Ailsa Craig granite
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	29 RUM_ZI3 1 2 54 32.19 6d bdl 0d bdl 18 0.23 8d 0.23	RUM_Z3b 1	RUM_Z12	RUM_Z21	RUM_ZB10	RUM_ZB2	RUM_ZB8	L GDD	TD11	13138_Z7	13138_Z3	13138_Z8	TAVA4070 E24 1
SiO ₂ (wt %) TTO ₂ A Al ₂ O ₃ MgO MgO MgO Al ₂ O ₃ Al ₂ O ₃ ThO ₂ ThO ₂	54 32.19 dd bdl bdl bdl bdl 18 0.23 dd 0.23 bdl bdl bdl		7	1	5	- 1	1	RUM_ZB9 1		5	2	2	U V 24820_E34.1 1
HO2 HO2 H20 H20 H20 H20 H20 H40 H40 H40 H40 H40 H40 H40 H40 H40 H4	dl bdl dd bdl 18 0.23 bdl bdl	32.04	32.01	31.57	31.98	32.23	29.13	29.18	29.50	32.04	31.63	31.88	32.25
Al ₂ O ₃ Al ₂ O ₃ Mg0 0.0 Mg2 0.0 <t< td=""><td>dl bdl 18 0.23 dl bdl</td><td>bdl</td><td>lbd</td><td>lbd</td><td>lbdl</td><td>0.16</td><td>0.13</td><td>0.14</td><td>0.24</td><td>lbd</td><td>0.04</td><td>0.05</td><td>0.06</td></t<>	dl bdl 18 0.23 dl bdl	bdl	lbd	lbd	lbdl	0.16	0.13	0.14	0.24	lbd	0.04	0.05	0.06
Тео* Тео* Сао Сао Сао Сао Тоо Тоо Тоо Тоо Тоо Тоо Тоо Т	18 0.23 bdl bdl	lbd	0.07	lbd	0.02	1.11	1.15	1.60	0.95	lbd	lbd	lbd	lbd
MgO CaO CaO CaO CaO CaO ThO2 ThO2 ThO2 ThO2 ThO2 ThO2 ThO2 ThO	dl bdl	0.09	0.45	09.0	0.11	0.53	0.45	1.33	0.46	0.11	0.10	0.07	0.02
CaO 2702 2702 2702 2702 260 11,1 1102 11,2 203 260 203 27,2 203 27,2 203 203 203 203 203 203 203 203 203 20		lbd	lbd	lbd	lbd	lbd	0.12	0.09	lpq	lbd	lbd	lþd	bdl
ZrO2 2rO2 1100 1002 1100 1002 1100 1002 1100 1002 1100 1002 1100 1002 1100 1002 1002 1002 1003 1000 1000	04 0.04	0.02	lbd	0.04	0.02	0.42	0.17	0.14	0.42	0.02	lbd	lbd	lbd
Hf02 Th02 Nb205 P205 Y203 S ² Y203 S ² Ce203 O ² O ² O ² O ² O ² O ² O ³ O ³ O ³ O ³ O ³ O ³ O ³ O ³	05 59.95	62.67	60.40	59.60	61.71	58.99	58.24	56.87	61.89	61.26	59.03	62.60	63.24
Tho ² 1002 1002 1002 102 102 102 102 102 102	28 0.76	1.77	0.58	1.34	1.27	1.06	1.00	0.70	1.40	1.44	1.29	1.31	0.74
UO2 P2O5 P2O5 P2O5 P2O5 P2O3 P2O3 P2O3 P2O3 P2O3 P2O3 P2O3 P2O3	33 0.52	0.62	0.64	2.21	1.45	lbd	0.32	0.28	lbd	0.53	1.18	0.41	0.21
Nb2 05 Y2 05 Y2 05 S2 03 S2 03 Nd2 03 Nd2 03 D1 D2 D3 D2 D3 D2 D3 D2 D3 D2 D3 D2 D3 D3 D3 D3 D3 D3 D3 D3 D3 D3	37 0.34	0.38	0.42	0.70	0.48	0.16	0.29	0.21	0.14	0.21	0.46	0.19	0.27
P205 Y203 X203 C L2203 C L2203 C L2203 0 0 0 B 203 C 03 C 03 C 03 C 03 C 03 C 03 C 03 C	lbd lbd	0.12	lbd	lbd	0.08	0.14	0.08	0.20	0.66	lþd	lbd	lbd	lbd
Y ₂ O ₃ La ₂ O ₃ Ce ₂ O ₃ P ₁₂ O ₃ B ₁₂ O ₃ D ₁ B ₁₂ O ₃ D ₁ B ₁₂ O ₃ D ₁ Co ₂ O ₃ D ₁ Co ₂ O ₃ D ₁ D ₁ Co ₂ O ₃ D ₁ D ₁ Co ₂ O ₃ D ₁ Co ₂ O ₃ Co ₂ Co ₃ CO ₃ CO ₂ CO ₃ CO ₃ CO ₃ CO ₂ CO ₃	46 0.20	0.15	0.26	0.32	0.19	0.25	0.38	0.40	0.17	0.37	0.44	0.27	0.35
La ₂ O ₃ Ce ₂ O ₃ Pr ₂ O ₃ Bu ₂ O ₃ Cd ₂ O ₃ Cd ₂ O ₃ Cd ₂ O ₃	76 3.66	1.71	2.89	2.70	2.03	2.32	4.41	5.32	1.77	1.97	2.37	1.46	1.90
Ce ₂ O ₃ b Pr ₂ O ₃ 0. Nd ₂ O ₃ 0.0. Eu ₂ O ₃ b Cd ₂ O ₃ b Cd ₂ O ₃ b	11 bdl	bdl	lbd	lbd	lþd	lbd	0.26	lbdl	lbd	0.13	lbd	lþd	lbd
Pr ₂ O ₃ Nd ₂ O ₃ Eu ₂ O ₃ Gd ₂ O ₃ b	dl 0.17	lbd	0.10	0.16	lþd	0.18	0.11	0.19	0.25	lþd	lþd	lbd	lbd
Nd ₂ O ₃ 0.0 Eu ₂ O ₃ b Gd ₂ O ₃ b	17 bdl	0.21	lbd	lbd	lbd	lbd	lbd	0.10	0.10	lþd	lbd	lbd	lbd
Eu ₂ O ₃ b Gd ₂ O ₃ b	01 bdl	lbd	0.18	lbd	lbd	0.11	lbd	0.13	0.14	lþd	lþd	lbd	lbd
Gd2O3 b	lbd lb	lbd	lbd	lbd	lbd	0.15	0.24	0.29	lbd	lbd	lbd	lbd	na
, (, (dl 0.15	0.14	lbd	0.12	0.18	0.24	0.47	0.15	lbd	0.26	0.18	lbd	lbd
DV,O, 0.	49 0.58	0.28	0.22	0.50	lbdl	0.48	0.23	0.29	lbd	lpd	lbd	lbd	lbd
Eroo ₃ 0.	75 0.33	0.13	0.30	0.30	0.23	0.29	0.38	0.62	0.21	0.17	0.38	0.24	0.41
Yb2O3 0.5	91 0.61	0.37	0.57	0.59	0.40	0.34	0.50	0.73	0.22	0.40	0.48	0.36	0.42
Total 99 .	43 99.72	100.68	99.07	100.74	100.14	99.15	98.08	98.96	98.52	98.80	97.56	98.83	99.87
												-	
Zr/Hf w/w 35	8.2 68.9	30.9	91.1	38.8	42.6	48.6	50.9	71.2	38.6	37.1	40.1	41.7	74.6
Formulae on the basis of	of 16 oxygens												
Si 3.9	98 4.018	3.968	4.008	3.961	3.984	3.991	3.747	3.719	3.740	4.006	4.021	3.985	3.978
I	1	I	I	I	I	0.015	0.013	0.013	0.023	I	0.004	0.004	0.006
N N	I	I	0.010	I	0.002	0.161	0.174	0.240	0.142	I			
Fe ²⁺ 0.0	19 0.024	0.009	0.047	0.063	0.011	0.055	0.048	0.142	0.049	0.011	0.010	0.007	0.002
Mg	1	I	I	I	I	I	0.024	0.017	I	I	I	I	I
Ca 0.00	05 0.006	0.002	I	0.005	0.002	0.056	0.023	0.020	0.057	0.002	I	I	I
Zr 3.4(55 3.649	3.785	3.689	3.647	3.748	3.562	3.653	3.535	3.827	3.736	3.660	3.815	3.804
Hf 0.0	46 0.027	0.063	0.021	0.048	0.045	0.037	0.037	0.025	0.051	0.051	0.047	0.047	0.026
Th 0.0	38 0.015	0.017	0.018	0.063	0.041	I	0.009	0.008	I	0.015	0.034	0.012	0.006
U 0.0	10 0.010	0.011	0.012	0.019	0.013	0.004	0.008	0.006	0.004	0.006	0.013	0.005	0.007
Nb	I	0.006	I	I	0.005	0.008	0.005	0.012	0.038	I	I	I	I
P 0.0	49 0.021	0.016	0.027	0.034	0.020	0.026	0.041	0.043	0.018	0.039	0.047	0.028	0.036
Y 0.3	88 0.243	0.113	0.192	0.180	0.134	0.153	0.302	0.360	0.119	0.131	0.160	0.097	0.125
La 0.0	05 -	I	I	I	I	I	0.013	I	I	0.006	Ι	I	I
Ce	- 0.008	I	0.004	0.007	I	0.008	0.005	0.00	0.011	I	I	I	I
Pr 0.0	- 80	0.009	I	I	I	I	I	0.005	0.005	I	I	I	I
PN	1	I	0.008	I	I	0.005	I	0.006	0.006	1	I	I	I
Eu	I	I	I	I	I	0.006	0.010	0.013	I	I	I	I	I
Gd	- 0.006	0.006	I	0.005	0.007	0.010	0.020	0.006	I	0.011	0.008	I	I
Dy 0.0.	20 0.023	0.011	0.009	0.020	I	0.019	0.010	0.012	I	1	I	I	I
Er 0.0.	30 0.013	0.005	0.012	0.012	0.00	0.011	0.015	0.025	0.008	0.007	0.015	0.009	0.016
Yb 0.0.	35 0.023	0.014	0.022	0.023	0.015	0.013	0.020	0.028	0.008	0.015	0.019	0.014	0.016
Σcations 8.	12 8.08	8.03	8.08	8.09	8.04	8.14	8.18	8.24	8.11	8.04	8.04	8.02	8.02

8.02	8.08	8.02	8.04	8.13	8.04	8.03	8.01	8.01	8.04	Σcations
0.010	0.026	0.013	0.016	0.022	0.004	0.018	0.007	0.006	0.006	<u>-</u> Үb
0.006	0.008	0.007	0.013	0.018	I	0.008	0.006	0.004	0.004	Er
I	I	I	0.009	0.008	I	I	I	I	1	Dy
I	0.007	0.005	0.005	0.007	I	0.006	I	I	0.010	Gd
I	0.006	I	I	I	I	I	I	I	I	Sm
I	I	I	0.007	1	I	I	I	0.006	I	Pr
	1	1	0.005	0.007	1		1			Ce
0.085	0.141	0.043	0.151	0.224	0.046	0.080	0.031	0.019	0.024	Y
0.026	0.039	0.026	0.057	0.043	0.022	0.047	0.009	0.010	0.007	P
	1		0.007	0.024	1	1	1 0	1	1	Nh (
0.005	0.037	0.011	0.010	0.006	0.004	0.007	0.008	0.004	0.005	U
0.010	0.065	0.011	0.026	0.036	0.003	0.006	0.008	0.002	0.005	Th
0.057	0.052	0.076	0.065	0.084	0.082	0.050	0.086	0.110	0.089	Hf
3.785	3.533	3.779	3.662	3.508	3.834	3.856	3.845	3.803	3.796	Zr
I	0.028	0.007	I	0.023	0.021	I	I	0.003	0.013	C_a
I	I	I	I	0.006	I	I	I	Ι	1	Mg
0.013	0.037	0.017	0.015	0.120	0.049	0.036	0.004	0.010	0.056	Fe^{2+}
0.012	0.062	0.012	0.007	0.007	I	I	I	0.003	1	AI
0.005	I	I	I	I	I	I	I	I	0.004	Ţ
4.010	4.038	4.014	3.986	3.984	3.974	3.917	4.003	4.032	4.022	Si
								/gens	the basis of 16 oxy	Formulae on
34.0	34.6	25.5	28.9	21.3	23.8	39.5	22.8	17.6	21.84	Zr/Hf w/w
99.55	99.05	99.52	99.14	99.93	98.45	101.04	99.87	99.54	99.93	Total
0.26	0.67	0.35	0.42	0.58	0.10	0.48	0.18	0.17	0.15	Yb ₂ O ₃
0.16	0.21	0.19	0.34	0.46	bdl	0.20	0.15	0.10	0.11	Er_2O_3
bdl	bdl	bdl	0.22	0.19	bdl	bdl	bdl	bdl	bdl	Dy_2O_3
bdl	0.17	0.12	0.12	0.17	bdl	0.16	bdl	bdl	0.24	Gd_2O_3
bdl	0.14	bdl	bdl	bdl	bdl	bdl	bdl	bdI	bdl	Sm_2O_3
bdl	bdl	bdl	0.16	bdl	bdl	bdl	bdl	0.12	bdl	Pr_2O_3
bdl	bdl	bdl	0.12	0.14	bdl	bdl	bdl	bdl	bdl	$Ce_{2}O_{3}$
1.30	2.11	0.66	2.27	3.36	0.69	1.23	0.47	0.28	0.37	$\tilde{Y_{2}O_{3}}$
0.25	0.36	0.25	0.54	0.40	0.20	0.45	0.09	0.10	0.07	P,O,5
bdl	bdl	bdl	0.13	0.42	bdl	bdl	bdl	bdl	bdl	Nb ₂ O ₅
0.18	1.33	0.40	0.34	0.22	0.15	0.27	0.28	0.13	0.17	UO,
0.35	2.25	0.38	0.90	1.28	0.10	0.23	0.27	0.07	0.19	ThO,
1.61	1.45	2.14	1.81	2.35	2.30	1.42	2.44	3.11	2.52	HfO_{2}
92 76	57 40	62 42	20 83 DOI	57 76	62 63	64 36	63 64	62 84	62.03	ZrO
DUI			DOT 1	0.03	0 16	Pd1	bdi Duli	0.02	01.0	
0.12	0.35	0.16	0.15	0.02	0.47	0.35	0.03	0.09	0.55	HeUT
0.08	0.42	0.08	0.05	0.05	bdl	DdI	bdl	0.02	bdl	AI_2U_3
0.05	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.04	TiO_2
32.42	31.99	32.33	31.76	31.71	31.66	31.89	32.31	32.49	32.52	SiO_2 (wt %)
4	2	4	2	2	2	3	2	з	2	N =
Grigadale Gran. 13138 Z2	aig Granite GLE2_Z7brim	GLE2 Z2	Maol Gain. 14709 Z1	Loch Ainort H1554 Z1c	ic Felsite NPFx1 Z4	N. Porphyriti NPFx1_Z1rim	ranite H1226 Z5rim	g Strollamus G H1226 Z3	Creag H1226 Z1rim	Intrusion Analysis
			Hills Centre	Western Red			entre	rn Red Hills C	Easte	
Ardnamurchan					Skye					
	_									

550

 Table 8. Representative hydrothermal type 2 zircon compositions.



Figure 5. Variation in Y_2O_3 versus ΣREE (wt %) in the studied zircons: (a) magmatic, (b) late-stage magmatic, (c) hydrothermal type 1, and (d) hydrothermal type 2.



Figure 6. Variation in ThO₂ versus UO₂ (wt %) in the studied zircons: (a) magmatic, (b) late-stage magmatic, (c) hydrothermal type 1, and (d) hydrothermal type 2.

6 Results – modes of occurrence and compositions of other Zr-bearing minerals

6.1 Baddeleyite

Baddeleyite was identified in two samples from Ardnamurchan (Granophyre, Centre 2, sample 14724, and the Grigadale Granophyre, 13138) and in the Rum Western Granite (RUM). In all three, it is abundant – more than 50 crystals per section. Baddeleyite occurs as thin isolated blades (Fig. 10a, b), as radiating clusters (Fig. 10c), and as residual crystals from baddeleyite \rightarrow zircon reactions (Fig. 10d). It also occurs, but less frequently, as a site for zirconolite precipitation (Fig. 10e, g). The majority of crystals in all three samples are small, typically 1–3 µm wide blades, and enclosed mainly in quartz (Fig. 10a) and feldspar (Fig. 10b).

The Rum and Ardnamurchan samples have textures that demonstrate various stages of baddeleyite reaction to zircon, from small zircon patches in baddeleyite (Fig. 10c) to residual baddeleyite amongst zircon (Fig. 10d). One case shows



Figure 7. Late-stage magmatic zircons, BSE images except where noted. (a) A $200 \,\mu\text{m}$ zircon (Zrn) filling an irregular space among quartz (Qz), albite (Ab), and K-feldspar (Kfs), with silicate-melt inclusions (MI). Mull, Toll Doire Granophyre, sample 44858. (b) A CL image of (a) showing typical magmatic zoning. The arrow shows the more CL-active core of a quartz grain. (c) A $100 \,\mu\text{m}$ zircon (Zrn) filling an irregular space in amphibole (Amp), silicate-melt inclusions (MIs), and apatite (Ap). Skye, Western Red Hills Centre, Loch Ainort Granite, sample H1554.



Figure 8. Hydrothermal type 1 zircons, BSE images except where noted. (a) A 450 μ m zircon (Zrn) filling a space among amphibole (Amp), albite (Ab), and K-feldspar (Kfs). The zircon grew around a partially altered euhedral chevkinite-(Ce) (Chv). Skye, Strath na Creitheach Centre, Ruadh Stac Granite, sample RS1. (b) A CL image of (a) showing a mottled pattern. (c) A 250 μ m zircon (Zrn) filling a space between quartz (Qz), K-feldspar, and albite (Kfs/Ab). Skye, Strath na Creitheach Centre, Ruadh Stac Granite, sample RS1. (d) A 12 μ m euhedral zircon (Zrn) in a triangular cavity partially filled with chlorite (Chl). The cavity is formed by inclusion-rich albite (Ab) and K-feldspar with quartz (Qz) and titanite (Ttn). Rum, Western Granite, sample RUM.

the remains of a zirconolite–baddeleyite pair in a late-stage zircon (Fig. 10e). The zircon composition and texture show it to be of late-stage hydrothermal origin.

Their small size, very thin habit, and common rounded surface from polishing rendered most EPMA experiments invalid due to the excitation volume interacting with host phases, secondary fluorescence, and instrument geometry problems. For example, Lumpkin (1999) suggests that many reported elements, such as Si, Al, and Ca, may be overestimated from electron beam–host interaction, and we cannot preclude that some of our element data result from such an interaction. There are significant concentrations in the UK baddeleyite of Nb₂O₅ (≤ 2.95 wt%) and TiO₂ (≤ 1.96 wt%) (Table 10). The presence of Nb⁵⁺ and Ta⁵⁺ requires a charge balanced substitution, such as $3Zr \leftrightarrow 2(Nb,Ta) + (Mg,Fe)$ (Lumpkin, 1999), as with the other identified 2⁺ and 3⁺ elements. A limited number of REEs were determined here, and all values were < 1 wt% and many at the detection limit, hence having a greater uncertainty than those elements > 1 wt%. Other instrumentation, such as laser ablation (LA) ICP-MS would yield better detection limits, although NanoSIMS may be the instrument of choice considering their



Figure 9. Hydrothermal type 2 zircons, BSE images except where noted. (a) A 175 μ m zircon (Zrn) filling a space between magnetite (Mag) and vugs (V) that shows the transition between type 1 (pt. 1) and type 2 (pt. 2) hydrothermal zircon. Skye, Eastern Red Hills Centre, Creag Strollamus Granite, sample H1226. (b) A 120 μ m euhedral zircon (Zrn) composed of an outer hydrothermal type 2 portion, filled with inclusions (e.g., calcite, REE+Y fluorcarbonate, xenotime, thorite, chlorite) around a euhedral magmatic zircon. The two crystals are inclined ~ 50° (solid lines) and enclosed by quartz (Qz) and magnetite (Mag). Skye, Western Red Hills Centre, (Northern) Porphyritic Felsite, sample NPF (sheared) #1. (c) CL image of (b) showing a cathodoluminescence-zoned magmatic center with a weak or non-CL hydrothermal portion. Barely visible bright spots in the outer portion are thorite and REE+Y fluorcarbonate. (d1,2,3) Inclusions in hydrothermal type 2 zircon (Zrn): (d1) thorite (Thr) inclusion from sample NPF sheared #1, void (V) space, SE image; (d2) inclusion composed of fergusonite (Fgs), thorite (Thr), and probable hingganite-(Y), sample H1226; and (d3) inclusion composed of three electron beam-sensitive crystals of an unknown U,Y,Si carbonate, dark areas are void (V) space, sample H1226. (e) A group of inclusion-filled zircons (Zrn) at a quartz(Qz) and K-feldspar–albite (Kfs/Ab) boundary, and many inclusions contain thorite. Skye, Western Red Hills Centre, Glamaig Granite, sample GLE2. (f) A 100 µm inclusion-filled zircon (Zrn) that formed in a gap in K-feldspar (Kfs) and albite (Ab). Skye, Western Red Hills Centre, Maol na Gainmich Granite, sample 14709. (g) A 30 µm zircon (Zrn) filling a vug (V) after the precipitation of fluornatropyrochlore (Pcl). The cavity in the zircon has a botryoidal surface (black arrows), as well as the upper portion (white arrow). The Zrn and Pcl are enclosed by K-feldspar (Kfs) and albite (Ab) accompanied by riebeckite (Rbk). Ailsa Craig microgranite, sample DV24828.

sizes observed in this study. Although most baddeleyite compositional studies in the literature do not include REEs, a few have reported REEs using either EPMA (e.g., Rajesh et al., 2006; Tropper et al., 2007; Huraiová et al., 2011; Zhao et al., 2016) or LA-ICP-MS (e.g., Schärer et al., 2011; Zhang et al., 2017) techniques. In all cases, the Σ REE content is very minor, ≤ 1 wt %–1.5 wt %.

6.2 Zirconolite

Zirconolite, ideally CaZrTi₂O₇, was identified only in the Ardnamurchan and Glamaig samples. The Glamaig zirconolite was analyzed by EPMA during a study focused on chevkinite-group minerals (Macdonald et al., 2013a). No description of the texture or petrographic relations of the GLE2 occurrence is available; nevertheless, we include two analyses in Table 11. In the Ardnamurchan samples, zirconolite forms small, $\leq 10 \,\mu$ m, anhedral to subhedral crystals frequently intergrown with other phases, such as britholite (Fig. 10f), baddeleyite (Fig. 10g), and zircon (Fig. 10h). Most crystals are $\leq 5 \,\mu$ m. Zirconolite in the Ardnamurchan samples is weakly CL active, and there was no zoning observed in either CL or BSE imaging.

Three polytypes of zirconolite have been described: zirconolite-2M, a monoclinic two-layered structure, zirconolite-3O, an orthorhombic three-layered structure, and

Table 9. Hydrothermal type 2 zircon compositions from Ailsa Craig microgranite.

			Ails	sa Craig sa	mple DV24	828		
Analysis	F716.1	F725 1	F735 1	F736 1	F736.2	F737 1	F738 1	F7391
N =	1	1	1	1	1	1	1	1
	1	1	1	1	1	1	1	
SiO ₂ (wt %)	32.20	29.93	29.81	30.79	30.57	28.16	31.06	32.22
110 ₂	0.42	0.71	0.92	0.57	0.52	0.68	0.35	0.40
Al_2O_3	1.15	1./8	1.21	1.28	1.44	1.99	1.13	1.27
FeO [*]	1.54	1.10	1.16	1.42	1.32	0.58	1.17	1.08
MaO	0.11 bdl	0.07 hdl	0.17 bdl	0.09 hdl	0.04	0.02	0.12 hdl	0.09 hdl
MgO CaO	2 37	0.00	2.05	2.82	2.47	1.00	3 22	2 94
KaO	0.03	0.03	bdl	bdl	0.18	0.08	0.20	0.13
ZrO2	47.65	45.17	44 43	49.52	46 78	45 79	39.84	39.85
HfO ₂	1.14	1.16	0.95	1.00	1.06	1.26	0.91	1.06
ThO ₂	bdl	bdl	0.24	bdl	0.36	bdl	0.07	0.33
UO ₂	bdl	bdl	0.33	0.45	0.31	0.15	0.33	0.33
Nb ₂ O ₅	0.41	1.00	1.53	0.89	0.72	0.88	2.56	2.33
Ta ₂ O ₅	0.22	bdl	bdl	0.38	0.22	bdl	0.09	0.30
P_2O_5	0.18	3.50	1.29	0.21	0.39	3.17	0.09	0.02
Y_2O_3	3.26	4.42	5.34	3.81	4.73	4.18	4.98	6.20
La ₂ O ₃	0.60	0.59	0.17	0.18	0.18	0.38	0.77	0.55
Ce_2O_3	1.11	1.68	0.71	0.43	0.53	1.84	1.16	1.19
Pr_2O_3	0.10	0.46	0.10	bdl	bdl	0.25	0.17	0.10
Nd ₂ O ₃	0.44	0.84	0.62	0.29	bdl	0.68	0.74	1.01
Sm_2O_3	0.29	0.38	0.12	bdl	0.28	0.28	0.16	0.49
Ga_2O_3	0.12	0.19	0.73	0.13	0.05	0.29	0.70	0.47
Dy_2O_3	0.49	0.47	0.99	0.01	0.84	0.39	0.89	0.95
$E_{12}O_3$ VbaOa	0.41	0.58	0.45	0.45	0.38	0.40	0.78	0.03
Cl	0.11	0.14	bdl	0.23 bdl	0.30	0.27	bdl	0.54 bdl
-0=Cl	0.20	0.07	-	-	0.00	0.17	-	_
Total	94.30	95.11	93.75	95.51	94.94	93.17	91.93	94.29
$\frac{7}{2r/Hf} w/w$	36.4	34.1	41.0	43.4	38.5	31.7	38.2	32.8
Formulae on t	the basis of	16 oxyger	15					
Si	4 167	3 808	3 924	3 971	3 980	3 690	4 178	4 220
Ti	0.041	0.068	0.091	0.056	0.051	0.067	0.035	0.040
Al	0.175	0.267	0.187	0.194	0.221	0.307	0.178	0.196
Fe ²⁺	0.145	0.123	0.127	0.153	0.144	0.063	0.131	0.118
Mn	0.012	0.008	0.019	0.009	0.005	0.002	0.014	0.010
Mg	-	-	_	-	0.079	0.046	-	-
Ca	0.329	0.135	0.289	0.390	0.345	0.152	0.464	0.413
Κ	0.004	0.005	_	-	0.029	0.013	0.034	0.021
Zr	3.007	2.803	2.852	3.115	2.970	2.926	2.613	2.545
Hf	0.042	0.042	0.036	0.037	0.039	0.047	0.035	0.040
Th	-	-	0.007	-	0.011	-	0.002	0.010
U	-	-	0.010	0.013	0.009	0.004	0.010	0.010
Nb To	0.024	0.057	0.091	0.052	0.042	0.052	0.156	0.138
1a D	0.008	0.277	0 1 4 4	0.013	0.008	0.252	0.003	0.011
r V	0.019	0.377	0.144	0.023	0.043	0.352	0.010	0.002
1 La	0.224	0.299	0.574	0.201	0.527	0.291	0.550	0.432
La Се	0.029	0.028	0.008	0.008	0.009	0.018	0.058	0.020
Pr	0.052	0.078	0.034	0.020	0.025	0.000	0.037	0.057
Nd	0.000	0.021	0.005	0.013	_	0.012	0.000	0.005
Sm	0.013	0.017	0.005		0.012	0.013	0.007	0.022
Gd	0.005	0.008	0.032	0.006	0.028	0.013	0.031	0.021
Dy	0.020	0.019	0.042	0.025	0.035	0.016	0.038	0.040
Er	0.017	0.015	0.018	0.017	0.024	0.016	0.033	0.027
Yb	0.004	0.006	0.019	0.009	0.012	0.011	0.018	0.014
Cl	0.044	0.014	-	_	0.017	0.038	-	
Total	8.41	8.24	8.34	8.39	8.47	8.27	8.49	8.46

 $FeO^* = total iron as Fe^{2+}; bdl = below detection limit; N = number of analyses averaged. PbO, SrO = not analyzed.$



Figure 10. Baddeleyite and zirconolite, BSE images. (a) Two doubly terminated blades of baddeleyite (Bdy) enclosed in quartz (Qz). Rum, Western Granite, sample RUM. (b) A 50 μ m blade of baddeleyite (Bdy) imbedded in quartz (Qz), albite (Ab), and K-feldspar (Kfs). A small portion on the left side of the blade has been plucked out. Rum, Western Granite, sample RUM. (c) A 50 μ m baddeleyite (Bdy) blade, in part with radiating blades, in K-feldspar (Kfs) and chlorite (Chl). Zircon (Zrn) is replacing the baddeleyite on the lower-right and upper-left ends. Rum, Western Granite, sample RUM. (d) A residual mass of baddeleyite (Bdy) that has been replaced and overgrown by zircon (Zrn), enclosed by quartz (Qz) and albite (Ab). Rum, Western Granite, sample RUM. (e) Small \sim 1 μ m crystals of zirconolite (Zrc) growing on a 3 μ m baddeleyite (Bdy), both enclosed by a 17 μ m euhedral zircon (Zrn) enclosed by quartz (Qz), albite (Ab), and K-feldspar (Kfs). Ardnamurchan, Grigadale Granophyre, sample 13138. (f) A 6 μ m euhedral zirconolite (Zrc) with a coeval, at least on the rim, britholite-(Ce)(Bri). Surrounded by quartz (Qz) and K-feldspar (Kfs) with a zoned apatite (Ap) to the lower left. Ardnamurchan, Grigadale Granophyre, sample 14724. (h) A 10 μ m subhedral zirconolite (Zrc) that has partially grown over a 3 μ m zircon (Zrn), enclosed by quartz (Qz) and albite (Ab). Ardnamurchan, Grigadale Granophyre, sample 14724. (h) A 10 μ m subhedral zirconolite (Zrc) that has partially grown over a 3 μ m zircon (Zrn), enclosed by quartz (Qz) and albite (Ab). Ardnamurchan, Grigadale Granophyre, sample 13138.

zirconolite-3T, a three-layered trigonal structure; two other polytypes, 4M and 6T, have been described as supercells of the 2M and 3T polytypes, respectively (for a detailed crystallographic discussion see Mazzi and Munno, 1983; White, 1984; Bayliss et al., 1989; Smith and Lumpkin, 1993; Coelho et al., 1997). We have no structural data and the crystal shapes do not suggest any particular polytype.

Although the nominal zirconolite formula is simple $(CaZrTi_2O_7)$, the composition in natural samples is complex and exhibits extensive element substitution as a function of both host rock and paragenesis (e.g., Williams and

Gieré, 1996; Gieré et al., 1998). In an attempt to classify and understand the various element substitution schemes, Gieré et al. (1998) defined five zirconolite end-members: CaZrTi₂O₇, CaZrMe⁵⁺Me³⁺O₇, REEZrTiMe³⁺O₇, REEZrMe⁵⁺Me²⁺O₇, and ACTZrTiMe²⁺O₇, where Me²⁺ = Mg²⁺ + Fe²⁺ + Mn²⁺ + Zn²⁺; Me³⁺ = Al³⁺ + Fe³⁺ + Cr³⁺; Me⁵⁺ = Nb⁵⁺ + Ta⁵⁺; REE = Ln³⁺ + Y³⁺; and ACT = U⁴⁺ + Th⁴⁺. Three structural sites can show extensive substitution: M8 = Σ (REE+Y)+ Σ (U+Th)+Ca; M5,6 = Σ Mg+Fe^{total}+Al+ Σ (Nb+Ta)+W+Ti; and M7 = Σ (Zr+Hf). However, Gieré et al. (1998) pointed out

	Table	10.	Comp	ositions	of	baddelevite.
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	Ardnamurchan					Rum					
Intrusion		Grigadale	Granophyre		Grano. Cen. 2			Western	Granite		
Analysis	13138 B4	13138 B2a	13138 B2b	13138 B2c	14724 B4	RUM B2a	RUM B2b	RUM B2.2	RUM B1a	RUM B7a	RUM B7b
N =	1	1	1	1	1	2	4	1	1	1	1
SiO ₂ (wt %)	bdl	bdl	0.04	bdl	1.02	0.40	0.31	0.39	0.33	0.93	bdl
TiO ₂	0.78	0.35	0.69	0.40	0.47	1.92	1.96	1.96	0.76	1.05	1.00
Al ₂ O ₃	bdl	bdl	bdl	bdl	0.25	0.09	0.03	0.13	0.44	0.96	0.53
FeO*	0.51	0.12	0.20	0.17	3.07	0.84	0.79	0.95	0.36	3.65	2.33
MgO	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.08	bdl
CaO	0.02	0.02	bdl	bdl	bdl	0.04	bdl	bdl	0.16	0.04	0.05
ZrO ₂	95.84	95.13	97.34	96.66	85.74	91.38	90.51	90.42	92.45	88.41	90.02
HfO ₂	1.31	3.47	1.08	1.06	4.08	1.58	1.62	1.71	1.74	1.75	1.80
ThO ₂	bdl	bdl	bdl	bdl	bdl	0.09	0.17	0.09	bdl	bdl	bdl
UO ₂	0.06	0.12	0.14	bdl	bdl	0.23	0.34	0.25	0.41	0.28	bdl
Nb ₂ O ₅	1.07	0.44	0.80	0.94	2.07	2.72	2.95	2.97	0.86	1.76	1.80
Y_2O_3	0.42	bdl	0.10	0.26	0.18	0.43	0.41	0.40	1.29	0.31	0.23
La_2O_3	bdl	bdl	bdl	0.12	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Ce_2O_3	0.13	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
Pr ₂ O ₃	bdl	bdl	bdl	0.24	bdl	0.12	0.10	0.25	bdl	bdl	bdl
Nd ₂ O ₃	0.24	bdl	bdl	bdl	bdl	0.16	0.10	0.13	0.10	bdl	bdl
Sm ₂ O ₃	bdl	0.17	bdl	bdl	bdl	bdl	bdl	bdl	0.23	bdl	bdl
Dy_2O_3	bdl	0.20	bdl	0.21	0.19	bdl	0.15	bdl	0.15	bdl	bdl
Er_2O_3	bdl	bdl	bdl	bdl	0.10	0.23	0.14	0.29	0.24	bdl	bdl
Yb_2O_3	0.15	bdl	bdl	bdl	bdl	0.17	0.22	0.11	0.17	bdl	0.35
Total	100.52	100.02	100.38	100.04	97.17	100.40	99.79	100.05	99.69	99.22	98.11
Formulae on t	he basis of fo	ur oxygens									
Si	-	-	0.002	-	0.043	0.016	0.013	0.016	0.014	0.038	-
Ti	0.024	0.011	0.021	0.012	0.015	0.058	0.060	0.060	0.024	0.032	0.031
Al		-	-	-	0.013	0.004	0.002	0.006	0.021	0.046	0.026
Fe ²⁺	0.017	0.004	0.007	0.006	0.109	0.028	0.027	0.032	0.012	0.124	0.081
Mg		-	-	-		-	-	-	-	0.005	-
Ca	0.001	0.001	-	-	0.001	0.002	_	-	0.007	0.002	0.002
Zr	1.911	1.925	1.938	1.937	1.769	1.805	1.804	1.794	1.862	1.757	1.827
Hf	0.015	0.041	0.013	0.012	0.049	0.018	0.019	0.020	0.021	0.020	0.021
Th		-	-	-		0.001	0.002	0.001	-	-	-
U	0.001	0.001	0.001	-		0.002	0.003	0.002	0.004	0.003	-
Nb	0.020	0.008	0.015	0.017	0.040	0.050	0.055	0.055	0.016	0.032	0.034
Y	0.009	-	0.002	0.006	0.004	0.009	0.009	0.009	0.028	0.007	0.005
La		-	-	0.002		-	-	-	-	-	-
Ce	0.002	-	-	-		-	-	-	-	-	-
Pr		-	-	0.004		0.002	0.001	0.004	-	-	-
Nd	0.003	-	-	-		0.002	0.001	0.002	0.001	-	_
Sm		0.002	-	-		-	_	-	0.003	-	_
Dy		0.003	-	0.003	0.003	-	0.002	-	0.002	-	_
Er		-	-	-	0.001	0.003	0.002	0.004	0.003	-	_
Yb	0.002	-	-	-	-	0.002	0.003	0.001	0.002	-	0.004
Σcations	2.01	2.00	2.00	2.00	2.05	2.00	2.00	2.01	2.02	2.07	2.03

 $FeO^* = total$ iron as Fe^{2+} ; bdl = below detection limit; N = number of analyses averaged. PbO, Eu₂O₃, Gd₂O₃ = bdl.

that although only the $CaZrTi_2O_7$ and $CaZrMe^{5+}Me^{3+}O_7$ end-members are found in nature, the others help define the various substitutions found in natural and synthetic zirconolites.

Given that most crystals in the granites are $\leq 5 \,\mu m$ and/or intergrown with other phases, successful microprobe analysis was difficult. Zirconolites from both localities show extensive, but similar, substitutions, although our data set is limited. Furthermore, not all the REEs or all the elements suggested by Gieré et al. (1998) were determined. CaO in ideal CaZrTi₂O₇ is 16.5 wt %, whereas in our study CaO varied from 2.90 wt %–5.34 wt % (Table 11). The deficiency in the M8 site is made up by REE+Y+U+Th. Figure 11a shows a good negative correlation between Ca and REE+Y (apfu) for Ardnamurchan and Glamaig zirconolites. The new analyses are, in fact, notably (REE+Y)-rich, with Y being dominant. Zirconolites with such high Y_2O_3 values (8.1 wt %–9.9 wt %) are uncommon (Williams and Gieré, 1996) but have been recorded in a sapphirine granulite of the Vestfold Hills, East Antarctica (Harley, 1994). The Glamaig phase is also notably Nb-rich (0.190–0.245 apfu).

The Glamaig zirconolites are enriched in U and Th relative to the Ardnamurchan zirconolite, and the M8 site sum for both localities varies from 0.90–0.97 apfu. Except for one sample, Y dominates the M8 site; hence these are zirconolite-(Y). Both the Ardnamurchan and Glamaig zirconolites have



Figure 11. Chemical variation in zirconolite in atoms per formula unit, based on seven oxygens. (a) Variation in REE+Y versus Ca. A linear regression line for the Ardnamurchan data is shown. A line connecting the Glamaig data is also shown that suggests a similar slope to the Ardnamurchan data. (b) Variation in Me^{5+} versus Ti. A linear regression line for the Ardnamurchan data is shown. A line connecting the Glamaig data is also shown that suggests a similar slope to the Glamaig data is also shown that suggests a similar slope to the Ardnamurchan data.

high total Fe and very low Mg and Mn. There is a good negative correlation between Me⁵⁺ and Ti (Fig. 11b), although we only have one Ta determination (Table 11). This correlation suggests that the substitution Me⁵⁺ + Me³⁺ \leftrightarrow 2Ti best describes the Me⁵⁺ relation and also suggests that most of the iron is ferric (Gieré et al. 1998).

The M8 site substitution for Ca is controlled by two factors: the ionic radius size and the abundance and diversity of potential cations in the environment of formation. A perusal of zirconolite databases such as Williams and Gieré (1996) shows the variability in REE+Y values in terrestrial and lunar zirconolites. They have low values of La^{3+} and Ce^{3+} compared to common LREE-enriched phases such as monazite or chevkinite. The eight-coordinated M8 site does not readily accept cations larger than Ca^{2+} (r = 0.112 nm) (Smith and Lumpkin, 1993). The larger ionic radii of La^{3+} (0.116 nm) and Ce³⁺ (0.114 nm) generally yield La- and Cepoor zirconolite relative to the other REE suite. A marked exception are zirconolites from xenoliths in the pyroclastic formations outcropping near Niedermendig, Laacher See eruptive center, Eifel volcanic region, Germany (Della Ventura et al., 2000) where zirconolite, baddeleyite, and monazite-(La) crystallized in voids from late-stage La-enriched metasomatic fluids.

Figure 12a shows chondrite-normalized REE patterns for the Ardnamurchan and Glamaig samples. The patterns reflect the low La and Ce substitution at the M8 site. All three samples are enriched in HREEs (Tb–Lu REE), and, when measured, the distribution shows a moderate Eu anomaly (Eu/Eu* = 0.34). Figure 12b plots four other HREE-enriched zirconolites: one from St. Kilda, Scotland, and three typical lunar zirconolites. Zirconolite, usually accompanied by baddeleyite, is a common accessory mineral in evolved lunar rocks (e.g., Wark et al., 1973; Roedder and Weiblen, 1978; Seddio et al., 2011; Norman and Nemchin, 2014). All the lunar examples come from late-stage mesostasis or residual liquids. The mesostasis composition typically varies from ~ 4 wt %–10 wt % alkalis and ~ 68 wt %–78 wt % SiO₂ (Potts et al., 2016). The lunar zirconolites also show strong HREE enrichment, with a large Eu anomaly.

The granite zirconolite analyses have low totals, 93.4 wt %-94.3 wt %. Although we did not analyze all the REEs or Ta – Ba in some analyses – and the iron may be mostly ferric, it is probable that part of the deficit is due to the presence of water, as previously reported in zirconolite by Zakrzewski et al. (1992).

6.3 Dalyite

Dalyite, the rare potassium zirconium silicate with only 14 known occurrences (Mindat, 2020), was identified only in the Ailsa Craig riebeckite microgranite where it is abundant; there are more than 25 crystals in the thin section. It occurs as a late-stage phase filling voids and cracks, although some appear to have crystal faces (Fig. 13a, b). It ranges in size from $\leq 5-30 \,\mu$ m, but most are $\leq 10 \,\mu$ m. Their small size precluded an optimal EPMA procedure, and some Na loss was noted. A common association is the occurrence of fluornatropyrochlore, a pyrochlore-group mineral (Atencio et al., 2010) that precipitated coevally with the dalyite (Fig. 13c, d). No other Zr-bearing phase is associated with dalyite formation, at least texturally. Dalyite is cathodoluminescent in blue, with very minor or no zoning.

The empirical formula of dalyite is $K_2ZrSi_6O_{15}$ (Back, 2018), but it can show limited substitutions. Jeffery et al. (2016) have summarized the possible site substitutions from a wide variety of lithological occurrences. The major exchange is Na \leftrightarrow K, with limited substitution in the tetrahedral and octahedral sites. The Ailsa Craig dalyite has mi-



Figure 12. (a) Chondrite-normalized plot of zirconolite from Ardnamurchan (samples 14724, 13138) and Glamaig (sample GLE2). Yttrium plotted for Ho. (b) Chondrite-normalized plot of zirconolite from 15065,93 Apollo 15 basalt (Wark et al., 1973), 12032,366-19 Apollo 12 granite (Seddio et al., 2011), St. Kilda pegmatite (Fowler and Williams, 1986), and 67955 Apollo 16 melt rock (Norman and Nemchin, 2014). The normalized Eu value for 67955 is given but not plotted. Normalizing factors from McDonough and Sun (1995).



Figure 13. Dalyite, BSE images from sample DV24828, Ailsa Craig microgranite. (a) A 25 μ m dalyite (Dly) crystal precipitated in an open space formed by quartz (Qz) and inclusion-rich K-feldspar–albite (Kfs/Ab). (b) A euhedral, partially fractured, 15 μ m dalyite (Dly) that grew on albite (Ab) in open space. (c) A 15 μ m dalyite (Dly) that filled an open space coeval with fluornatropyrochlore (Pcl). (d) A 15 μ m dalyite (Dly) filling an open space in K-feldspar and albite (Kfs/Ab) coeval with quartz (Qz) and fluornatropyrochlore (Pcl).



Figure 14. Variation diagram of Zr versus (Ti+Hf+Fe+Mg) apfu, showing substitution in the octahedral site in dalyite from sample DV24828, EPMA data, adapted from Jeffery et al. (2016). Also shown is the ideal dalyite composition (star) and dalyite compositional Group 1 (Terceira, Straumsvola, Ascension, Strange Lake) and Group 2 (Sunnfjord, Cancarix, Murun) from Jeffery et al. (2016).

nor substitution in both the K/Na polyhedra and the octahedral site ((Ti+Hf+Fe+Mn+Al) \leftrightarrow Zr) (Table 12). Figure 14 shows the octahedral site substitution and a comparison with other dalyite occurrences. All data plot within 0.05 apfu of the one-to-one line, indicating a one-to-one cation exchange in the site, although, with the low degree of substitution, there is only a minor davanite (K₂TiSi₆O₁₅) component. The K/Na site has a greater degree of substitution, with K+Na+Ba from 1.79–1.96 apfu. The lower values are the probable result of Na loss, with a concomitant increase in Si. The range of apfu is as follows: K 1.71–1.88, Na 0.028– 0.200, Ti 0.004-0.033, Hf 0.009-0.017, Al 0.003-0.101, and Fe 0.002–0.015; Mn (0.001–0.002 apfu) was detected in only 20 % of the experiments, and Mg was below detection. Barium, Ca, and P contents were minor or below detection. The dalyite analyses in the Ailsa Craig sample yield the average formula (K_{1.78}Na_{0.13})_{Σ1.91}(Zr_{0.94}Ti_{0.011}Hf_{0.013}Fe_{0.008} $Al_{0.019})_{\Sigma 0.991}Si_{6.04}O_{15}.$

The Ailsa Craig dalyite composition is similar to those in Group 1 of Jeffery et al. (2016) – low octahedral (Fig. 14) but greater K \leftrightarrow Na site substitution (Table 12) – which includes Terceira Island, Azores (Jeffery et al., 2016), Straumsvola Complex, Antarctica (Harris and Rickard, 1987), Strange Lake Complex, Canada (Birkett et al., 1992; Salvi and Williams-Jones, 1995), and the type locality, Ascension Island, South Atlantic Ocean (Van Tassel and Hey, 1952). Dalyite from these rocks reflects their low TiO₂ content and sodic character. The Ailsa Craig microgranite has low TiO_2 (0.11 wt %) and is sodic (Na/K 2.2 mol.) (Govindaraju, 1987; Table 2).

6.4 Eudialyte-group minerals

Eudialyte-group minerals (EGMs) are present in many forms in the Ailsa Craig microgranite, where it is abundant; there are more than 50 crystals in the thin section. EGMs are late phase and fill cracks and voids (Fig. 15a, b). They range in size up to $45 \,\mu\text{m}$, but most are $< 15 \,\mu\text{m}$. A euhedral crystal with two bladed offshoots and a partially rounded fluorite are enclosed in K-feldspar and albite (Fig. 15a). An aggregate of anhedral crystals is intergrown with quartz and feldspar in Fig. 15b with an intergrown britholite grain (Bri). The platy crystal in Fig. 15d has a dark core with a high Z mantle, and the prismatic crystal in Fig. 15e shows what may be sector zoning. A common association with EGMs is Zr-bearing aegirine and quartz (Fig. 15a, b, c, d). No intergrowths of dalyite, EGMs, or hydrothermal type 2 zircon were observed, although they are all late-stage phases. Zoning was observed in a few EGM crystals where a brighter BSE zone texturally follows a darker BSE zone separated by a sharp boundary (Fig. 15d, e).

Eudialyte-group minerals are a complex group of trigonal cyclosilicates with both three-fold (Si₃O₉)₆ and ninefold (Si₉O₂₇)₁₈ rings. A zeolite-like heteropolyhedral framework allows a considerable variety of elemental substitutions in different structural sites yielding a mineral group of structural complexity and chemical variability, with variable amounts of K, Sr, Fe, Mn, REEs, Y, Nb, Ti, and other elements (Johnsen and Gault, 1997; Johnsen and Grice, 1999; Johnsen et al., 2003; Bulakh and Petrov, 2004; Chukanov et al., 2004; Rastsvetaeva, 2007; Schilling et al., 2011; Rastsvetaeva et al., 2012; Borst et al., 2020). Chukanov et al. (2018) note that there are 28 members of the eudialyte group. The formula accepted by the International Mineralogical Association (IMA; Johnsen et al., 2003) for the group is $N_{15-16}[M1]_6[M2]_3[M3][M4]Z_3Si_{24}O_{66-72}]$ (Θ)₀₋₉ X_2 , where $N = Na^+$, Ca^{2+} , K^+ , Sr^{2+} , REE^{3+} , Y^{3+} , \Box (vacancy), Ba^{2+} , Mn^{2+} , or H_3O^+ ; $M1 = Ca^{2+}$, Mn^{2+} , REE^{3+} . Y^{3+} , Na⁺, Sr²⁺, or Fe²⁺; $M^2 = Fe^{2+}$, Mn²⁺, Na⁺, Zr⁴⁺, Ta^{5+} , Ti^{4+} , K^+ , Ba^{2+} , \Box , H_3O^+ , or Fe^{3+} ; $M3/4 = Si^{4+}$, Al³⁺, Nb⁵⁺, Ti⁴⁺, W⁶⁺, or Na⁺; $Z = Zr^{4+}$, Hf⁴⁺, Ti⁴⁺, or Nb⁵⁺; $\Theta = O^{2-}$, OH⁻, H₂O, CO₃²⁻, SO₄²⁻, or SiO₄⁴⁻; and $X = Cl^{-}$, F⁻, or OH⁻. Because of the site occupation complexity, the possible presence of different iron valences, vacancies, and hydrous species, the calculation of balanced mineral formulae is problematic, and this is especially true for EPMA data. The normalization scheme suggested by Johnsen and Grice (1999) is used here, in which the sum of (Si+Zr+Ti+Nb+Al+Hf+Ta) cations was normalized to 29 cations per formula unit, and all other cation values result from this normalization (Table 13).

Table 11. Compositions of zirconolite.

			Ardnamurchan			East. Red	Hills Centre
Intrusion	Grig Grano		Granophy	re Centre 2		Glamai	g Granite
Analysis N =	13138_ZLa 1	14724-ZL2 1	14724-ZL3 2	14724-ZL4 1	14724-ZL2b 1	GLE2a 1	GLE2b
SiO ₂ (wt %)	1.98	0.60	1.37	0.44	2.38	0.10	0.08
TiO ₂	26.78	27.34	23.14	23.40	26.58	23.07	21.43
Al ₂ O ₃	0.14	0.24	0.63	0.12	0.11	0.18	0.16
MnO	10.82	10.52	8.38	10.17	9.07	9.77	0.21
MgO	0.24	bdl	bdl	bdl	bdl	bdl	bdl
CaO	3.19	2.90	5.34	4.20	3.47	3.92	4.12
BaO	na	na	na	na	na	0.09	0.10
PbO	0.19	bdl	0.10	bdl	bdl	na	na
ZrO ₂ HfO ₂	23.22	27.56	27.98	27.44	27.95	28.55	27.55
ThO ₂	1.61	0.79	1.13	1.58	0.99	3.63	3.70
UO ₂	0.99	0.33	0.58	0.78	0.30	2.18	2.55
Nb ₂ O ₅	4.10	0.86	7.81	7.14	1.25	5.79	7.34
Ta ₂ O ₅	na	na	na	na	na	bdl	0.12
P ₂ O ₅	0.06	0.07	0.12	0.05	0.05	bdl	bdl
Y ₂ O ₃	0.10	0.12	7.97 0.46	0.45	10.03	9.93	9.30
Ce ₂ O ₃	1.03	2.12	2.58	2.65	2.03	1.66	1.66
Pr ₂ O ₃	0.24	0.40	0.40	0.11	0.41	0.27	bdl
Nd_2O_3	1.52	2.78	2.70	1.87	2.81	1.34	1.29
Sm_2O_3	0.83	0.81	0.96	0.50	1.07	0.34	0.32
Eu ₂ O ₃	0.14	bdl	bdl	0.16	0.14	bdl	bdl
Gd_2O_3	1.00	1.56	0.97	1.12	1.11	0.96 bdl	1.11 bdi
Dv_2O_3	1 52	1 49	1 29	0.85	1.86	1.00	1 17
H02O3	na	na	na	na	na	0.28	bdl
Er ₂ O ₃	1.49	1.21	0.78	0.99	1.19	na	na
Yb ₂ O ₃	1.22	0.76	0.69	1.01	0.56	1.03	1.10
Total	94.64	93.41	96.76	96.03	95.27	94.28	93.40
ΣREE $\Sigma REE + Y$	9.1 20.5	11.3 21.3	10.8 18.8	9.7 19.8	11.6 21.7	7.1 17.0	6.9 16.2
Formulae on t	the basis of sever	n oxygens					
Si	0.139	0.043	0.095	0.031	0.166	0.007	0.006
Ti	1.409	1.477	1.201	1.246	1.392	1.263	1.194
Al	0.012	0.021	0.051	0.010	0.009	0.015	0.014
Fe ²⁺	0.633	0.632	0.495	0.602	0.563	0.595	0.622
Mn Ma	0.025	_	0.001	_	_	0.014	0.013
Ca	0.0239	0.223	0.394	0.319	0.259	0.306	0.327
Ba	-	_	-	-	-	0.003	0.003
Pb	0.004	-	0.002	-	-	-	-
Zr	0.792	0.965	0.942	0.947	0.948	1.006	0.987
HI Th	0.016	0.020	0.023	0.018	0.017	- 0.060	0.062
U	0.020	0.015	0.013	0.023	0.010	0.000	0.002
Nb	0.130	0.028	0.244	0.229	0.039	0.191	0.246
Та	-	-	-	-	-	-	0.002
Р	0.004	0.004	0.007	0.003	0.003	-	-
Y	0.425	0.382	0.292	0.380	0.371	0.384	0.366
La	0.003	0.003	0.012	0.012	0.012	0.005	0.007
Dr Dr	0.026	0.036	0.065	0.009	0.032	0.044	0.045
Nd	0.038	0.071	0.066	0.047	0.070	0.035	0.034
Sm	0.020	0.020	0.023	0.012	0.026	0.009	0.008
Eu	0.003	-	_	0.004	0.003	-	-
Gd	0.023	0.037	0.022	0.026	0.026	0.023	0.027
Tb	-	-	-	-	-	-	-
Dy	0.034	0.034	0.029	0.019	0.042	0.023	0.028
Er	0.033	0.027	0.017	0.022	0.026	0.006	_
Yb	0.026	0.027	0.017	0.022	0.012	0.023	0.025
Σcations	4.08	4.09	4.03	4.06	4.06	4.05	4.06
ΣREE	0.213	0.276	0.258	0.236	0.277	0.175	0.175
$\Sigma REE + Y$	0.638	0.658	0.551	0.616	0.648	0.559	0.541
M8	0.92	0.90	0.97	0.97	0.93	0.96	0.97

 $\frac{100}{\text{FeO}^{*} = \text{total iron as Fe}^{2+}; N = \text{number of analyses averaged; bdl} = \text{below detection limit; na = not analyzed. SrO = bdl. M8 = <math>\Sigma \text{REE} + Y + U + \text{Th} + Ca.}$

Analysis	D21.1-2	D'/.1	D20.1	D3.1	D17.1-2	D23.1	D14.1-2	Range
N =	2	I	1	1	2	1	2	n = 15
SiO ₂ (wt %)	63.24	64.19	63.54	63.03	62.95	63.12	63.37	62.95-63.89
TiO ₂	0.27	0.25	0.34	0.06	0.46	0.06	0.07	0.05-0.46
Al ₂ O ₃	0.04	0.06	0.05	0.70	0.04	0.50	0.05	0.03-0.90
FeO*	0.06	0.09	0.07	0.10	0.14	0.11	0.07	0.02-0.19
MnO	bdl	bdl	bdl	0.02	bdl	bdl	0.02	bdl-0.02
MgO	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl
CaO	bdl	0.02	bdl	bdl	bdl	0.02	bdl	bdl-0.02
BaO	bdl	0.08	0.05	bdl	0.06	bdl	bdl	bdl-0.08
Na ₂ O	0.79	0.41	0.82	0.59	0.58	0.59	0.95	0.15-1.09
K ₂ O	14.40	14.26	14.06	15.13	14.26	14.67	14.64	14.06-15.52
ZrO ₂	20.00	20.70	20.09	20.34	19.96	20.25	20.52	19.96-20.93
HfO ₂	0.45	0.64	0.53	0.40	0.46	0.47	0.49	0.35-0.64
P_2O_5	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl-0.04
Total	99.24	100.26	99.55	100.37	98.91	99.78	100.16	98.91-100.59
Formulae on the basis of 1	5 oxygens							
Si	6.053	6.046	6.055	5.993	6.046	6.020	6.034	5.990-6.055
Ti	0.019	0.018	0.024	0.004	0.033	0.004	0.005	0.004-0.033
Al	0.005	0.007	0.005	0.078	0.004	0.056	0.005	0.004-0.101
Fe ²⁺	0.005	0.007	0.005	0.008	0.011	0.009	0.005	0.002-0.015
Mn	_	_	0.001	0.001	_	_	0.002	0.002
Mg	_	-	_	_	_	_	_	_
Ca	_	0.002	_	_	_	0.002	_	0.002
Ba	_	0.003	0.002	_	0.002	_	_	0.003
Na	0.146	0.076	0.151	0.109	0.107	0.108	0.175	0.028-0.200
Κ	1.758	1.725	1.709	1.835	1.747	1.785	1.777	1.709-1.884
Zr	0.933	0.957	0.933	0.943	0.935	0.942	0.953	0.930-0.971
Hf	0.012	0.017	0.015	0.011	0.013	0.013	0.013	0.009-0.017
Р	_	_	_	_	_	_	_	0.003
Σcations	8.93	8.86	8.90	8.98	8.90	8.94	8.97	8.86-8.98
$\Sigma(K+Na+Ba)$	1.90	1.80	1.86	1.94	1.86	1.89	1.95	1.80-1.96
Σ (Zr+Hf+Ti+Fe+Mn)	0.97	1.00	0.98	0.97	0.99	0.97	0.98	0.95 - 1.00

Table 12. Representative compositions of dalyite in Ailsa Craig granite DV24828.

 $FeO^* = total iron as Fe^{2+}$; bdl = below detection limit; N = number of analyses averaged.

Electron beam analysis was challenging because in most cases, a recommended $20 \times 20 \,\mu\text{m}$ defocused beam could not be used (Atanasova et al., 2015). The EGM in sample DV24828 was beam sensitive; Na and Cl were analyzed first, but Na loss and/or dehydration cannot be precluded. However, a comparison of the major element ranges (Table 13), Fe, Mn, Na, Ca, Sr, Si, Nb, Ti, Zr, and Cl, with the element ranges from a variety of EGM localities (e.g., Schilling et al., 2011; Table 3) shows that the Ailsa Craig EGMs are well within typical compositional ranges, except for Ca. In typical EGMs there are six Ca atoms per chemical formula, and these atoms are involved in a six-membered octahedral ring (e.g., Johnsen and Grice, 1999). The typical CaO content of IMA-accepted EGM species varies from 7 wt %-12 wt % (e.g., Bulakh and Petrov, 2004; Schilling et al., 2011). In oneillite, however, there are only three Ca atoms per formula unit (Johnsen et al., 1999), yielding a typical oneillite CaO content of 2.90 wt %–3.13 wt %. The CaO content of the Ailsa Craig EGM varies from 0.06 wt %–1.93 wt % (Table 13) with most < 0.50 wt % and only two grains > 1 wt % CaO. This measurement yields apfu that vary from 0.03–1.03 based on the recommended normalization. The EGM *M*1 site is usually filled by Ca but can accommodate other cations as per the substitution Ca \leftrightarrow (Mn, REEs, Y, Na, Sr, Fe).

We observed only a few zoned eudialyte grains, and only one was amenable to EPMA, although we obtained qualitative EDS data on others. The dark zone is richer in Na and Zr but poorer in Σ REE, with the lighter zone having the opposite relative abundance (Table 13). Harris et al. (1982) describe similar zoning in late-stage eudialyte in a dalyitebearing granite.

Ailsa Craig EGM is highly enriched in Y and REEs with some of the highest weight percentage values reported; Y_2O_3 ranges from 2.62 wt %–5.04 wt %, and Σ REE ranges from

Table 13. Representative compositions of probable eudialyte-group minerals in Ailsa Craig granite DV24828.

Analysis	E9.1-2	E17.1	E18.1	E21.1	E23.1	E26.1	E32.1	E33.1	E14.1	E14.2	Range
N =	2	1	1	1	1	1	1	1	dark 1	light 1	n = 21
	-			•				•	uun 1	ingin i	
SiO ₂ (wt %)	50.38	50.60	50.35	47.79	49.79	49.49	50.69	49.25	52.01	52.77	47.79–52.77
TiO ₂	0.40	0.42	0.30	0.68	0.66	0.38	0.50	0.55	0.28	0.38	0.28-0.68
Al_2O_3	bdl	bdl	bdl	bdl	bdl	bdl	bdl	bdl	0.03	0.03	bdl-0.03
FeO*	5.18	4.91	3.87	5.37	4.79	4.80	5.28	5.21	3.94	5.10	3.69-6.07
MnO	1.13	2.07	1.13	1.59	0.87	1.87	1.69	1.58	1.08	1.12	0.87 - 2.11
CaO	0.24	0.46	0.32	0.89	0.18	1.93	0.42	0.41	0.13	0.14	0.06-1.93
SrO	0.05	0.06	bdl	bdl	bdl	bdl	0.08	0.16	bdl	0.08	bdl-0.16
Na ₂ O	13.88	11.77	13.35	14.15	14.06	11.32	13.38	14.11	13.83	11.74	10.53-14.66
K ₂ O	0.48	0.40	0.47	0.39	0.40	0.37	0.34	0.36	0.16	0.28	016-0.55
ZrO_2	14.69	11.24	13.17	12.54	15.29	9.15	12.22	12.22	16.04	12.88	9.15-16.17
HfO_{2}	0.12	0.25	bdl	0.19	0.22	bdl	0.24	0.38	0.15	0.34	bdl-0.43
Nb2O5	1.17	2.11	2.05	1.07	1.13	2.31	1.89	1.40	1.80	2.08	0.93-2.31
Ta ₂ O ₅	0.10	bdl	0.32	0.12	bdl	0.19	0.10	bdl	bdl	bdl	bdl-0.32
P2O5	bdl	0.07	bdl	bdl	0.07	0.07	bdl	bdl	bdl	bdl	bdl-0.14
¥203	2.93	4.25	4.73	3.55	3.04	3.93	4.04	3.46	3.92	3.83	2.62 - 5.04
La2O2	0.73	1.12	0.90	1.01	0.80	0.98	1.22	0.81	0.58	1.17	0.58 - 1.22
Ce_2O_2	2.02	2.79	2.45	2.56	2.22	3.21	3.07	2.63	1.62	2.89	1.62-3.21
Pr_2O_3	0.26	0.20	0.16	0.16	0.41	0.39	0.41	0.91	bdl	0.39	bd1-0.91
Nd2O2	1.03	1.86	1 34	1.83	1.02	2.15	1.60	1 39	1.08	1 37	0.60-2.15
Sm2O2	0.12	0.64	0.53	0.44	0.22	0.42	bdl	bdl	bdl	bdl	bdl_0.68
Gd2O2	0.12	0.01	0.63	0.11	0.51	0.38	0 44	0.37	0.53	0.17	bdl_0.00
	0.48	0.80	0.55	0.15	0.35	0.98	0.67	0.88	0.33	0.78	0 15_0 98
$E_{r_2}O_2$	0.40	0.50	0.33	0.15	0.35	0.31	0.07	0.00	0.40	0.70	0.15 0.90
Zh ₂ O ₃	0.73	0.34	0.45	0.05	0.55	0.30	0.33	0.10	0.14	0.47	0.10 0.09
Cl	1.05	1 00	2.02	2.00	2.11	2.07	1.03	1.07	2.02	2 10	1 02_2 10
$O = C^1$	0.44	0.45	0.46	0.45	0.48	0.47	0.44	0.44	0.46	0.50	1.92-2.19
-0=01	0.44	0.45	0.40	0.+5	0.40	0.47	0.44	0.44	0.40	0.50	
Total	97.89	99.29	99.08	97.44	98.12	96.60	100.36	98.07	99.87	99.91	95.9–100.4
ΣREE	5.57	9.14	7.46	7.55	6.00	9.19	8.00	7.46	4.94	7.44	4.94-9.19
$\Sigma REE + Y$	8.49	13 39	12.19	11 10	9.04	13.12	12.04	10.92	8 86	11.27	7 91-13 39
		10.07	12.12	11110				10.72	0.00	11.27	1.71 15.57
<u> </u>	1 : 64	0	1.7.110		20			10.02	0.00	1112/	1.91 15.59
Cations on the	e basis of (Si+Ti+A	l+Zr+Hf-	+Nb+Ta)	= 29			10072	0.00		1.91 15.59
Cations on the	e basis of (25.001	Si+Ti+A 25.557	1+Zr+Hf- 25.171	+Nb+Ta)	= 29 24.761	25.939	25.355	25.339	24.754	25.332	24.67-25.94
Cations on the Si Ti	e basis of (25.001 0.151	Si+Ti+A 25.557 0.158	1+Zr+Hf- 25.171 0.111	+Nb+Ta) 25.204 0.269	= 29 24.761 0.247	25.939 0.148	25.355 0.188	25.339 0.214	24.754 0.099	25.332 0.137	24.67–25.94 0.10–0.27
Cations on the Si Ti Al	e basis of (25.001 0.151	Si+Ti+A 25.557 0.158	1+Zr+Hf- 25.171 0.111	+Nb+Ta) 25.204 0.269	= 29 24.761 0.247	25.939 0.148	25.355 0.188	25.339 0.214	24.754 0.099 0.017	25.332 0.137 0.018	24.67–25.94 0.10–0.27 – –0.02
Cations on the Si Ti Al Fe	e basis of (25.001 0.151 - 2.150	Si+Ti+A 25.557 0.158 - 2.074	1+Zr+Hf- 25.171 0.111 - 1.618	+Nb+Ta) 25.204 0.269 - 2.368	= 29 24.761 0.247 - 1.992	25.939 0.148 	25.355 0.188 	25.339 0.214 	24.754 0.099 0.017 1.568	25.332 0.137 0.018 2.047	24.67–25.94 0.10–0.27 – –0.02 1.55–2.58
Cations on the Si Ti Al Fe Mn	e basis of (25.001 0.151 - 2.150 0.475	Si+Ti+A 25.557 0.158 - 2.074 0.886	1+Zr+Hf- 25.171 0.111 - 1.618 0.480	+Nb+Ta) 25.204 0.269 - 2.368 0.710	= 29 24.761 0.247 - 1.992 0.365	25.939 0.148 2.104 0.830	25.355 0.188 - 2.209 0.716	25.339 0.214 - 2.242 0.689	24.754 0.099 0.017 1.568 0.435	25.332 0.137 0.018 2.047 0.454	24.67–25.94 0.10–0.27 – –0.02 1.55–2.58 0.36–0.94
Cations on the Si Ti Al Fe Mn Ca	e basis of (25.001 0.151 - 2.150 0.475 0.125	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505	= 29 24.761 0.247 $-$ 1.992 0.365 0.096	25.939 0.148 	25.355 0.188 2.209 0.716 0.222	25.339 0.214 - 2.242 0.689 0.225	24.754 0.099 0.017 1.568 0.435 0.067	25.332 0.137 0.018 2.047 0.454 0.070	$24.67-25.94 \\ 0.10-0.27 \\0.02 \\ 1.55-2.58 \\ 0.36-0.94 \\ 0.03-1.08$
Cations on the Si Ti Al Fe Mn Ca Sr	e basis of (25.001 0.151 2.150 0.475 0.125 0.013	Si+Ti+A 25.557 0.158 2.074 0.886 0.251 0.018	12.17) 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 -	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 -	= 29 24.761 0.247 1.992 0.365 0.096	25.939 0.148 2.104 0.830 1.084	25.355 0.188 2.209 0.716 0.222 0.022	25.339 0.214 - 2.242 0.689 0.225 0.049	24.754 0.099 0.017 1.568 0.435 0.067	25.332 0.137 0.018 2.047 0.454 0.070 0.023	$\begin{array}{r} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na	e basis of (25.001 0.151 2.150 0.475 0.125 0.013 13.356	Si+Ti+A 25.557 0.158 2.074 0.886 0.251 0.018 11.526	12.17 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558	25.939 0.148 	25.355 0.188 2.209 0.716 0.222 0.022 12.977	25.339 0.214 2.242 0.689 0.225 0.049 14.076	24.754 0.099 0.017 1.568 0.435 0.067 	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K	25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258	12.17 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260	= 29 24.761 0.247 $-$ 1.992 0.365 0.096 $-$ 13.558 0.251	25.939 0.148 2.104 0.830 1.084 - 11.504 0.247	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214	25.339 0.214 2.242 0.689 0.225 0.049 14.076 0.236	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr	25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768	12.17 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225	= 29 24.761 0.247 1.992 0.365 0.096 	25.939 0.148 2.104 0.830 1.084 	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066	24.754 0.099 0.017 1.568 0.435 0.067 	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K K Zr Hf	25.001 0.151 -2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035	12.17 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031	25.939 0.148 2.104 0.830 1.084 11.504 0.247 2.339	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035	25.339 0.214 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055	24.754 0.099 0.017 1.568 0.435 0.067 	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047	$\begin{array}{r} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb	25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.258 2.768 0.482	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256	= 29 24.761 0.247 	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 0.547	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0 387	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.045	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta	25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.203 0.013	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.482	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.226 0.028 0.028	= 29 24.761 0.247 	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.0547	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.025 0.427 0.013	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P	25.001 0.151 	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.482 - 0.030	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.027	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hf Nb Ta P Y	25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.482 - 0.030 0.035	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018 0.955	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.027 0.029 1.095	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 - - 0.946	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P Y La	2 basis of (25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772 0.133	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.482 - 0.030 1.141 0.208	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018 - 0.995 0.106	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147	25.939 0.148 - 2.104 0.830 1.084 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 - - 0.946 0.154	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - 0.977 0.207	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10.023\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P Y La Ca	2 basis of (25.001 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772 0.133 0.366	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.56 0.258 2.768 0.035 0.482 - 0.030 1.141 0.203 1.141	12.19 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448	11113 +Nb+Ta) 25.204 0.269	= 29 24.761 0.247 1.992 0.365 0.096 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.256	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 0.055 0.325 - - 0.946 0.154 0.405	24.754 0.099 0.017 1.568 0.435 0.067 - - 12.763 0.100 3.723 0.020 0.387 - - - 0.992 0.102 0.222	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28\\ 0.62\\ \end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P Y La Ce Pr	25.001 2.150 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772 0.133 0.366 0.475 0.017 0.263 0.013 - 0.772 0.133 0.365 0.072 0.133 0.365 0.072 0.133 0.365 0.072 0.133 0.151 - 0.151 - 0.151 - 0.151 - 0.151 - 0.155 0.127 0.263 0.013 - 0.133 0.126 0.125 0.123 0.133 0.136 0.047 0	Si+Ti+A 25.557 0.158	12.19 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.231	= 29 24.761 0.247 1.992 0.365 0.096 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.407 5.251 0.254 0.251 0.254 0.251 0.254 0.251 0.254 0.255	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.618	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.525	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 0.055 0.325 - - - 0.946 0.154 0.471	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - - - - - - - - - - - - - - - - - - -	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P Y La Ce Pr Nd	25.001 2.150 0.151 - 2.150 0.475 0.125 0.013 13.355 0.017 0.263 0.013 - 0.772 0.133 0.366 0.477 0.475 0.017 0.263 0.013 - 0.772 0.133 0.366 0.477 0.475 0.017 0.263 0.017 0.013 0.017 0.0263 0.017 0.013 0.017 0.0263 0.017 0.017 0.0263 0.027 0.0263 0.027 0.027 0.0263 0.027 0.0263 0.027 0.0263 0.027 0.0263 0.0266 0.0266 0.0266 0.027 0.0263 0.0266 0.0266 0.0266 0.0266 0.027 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0263 0.0266 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.0272 0.00772 0.0272 0.0	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.253 0.482 - 0.030 1.141 0.208 0.515 0.325	12.19 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.230	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.265 0.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.344	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.404 0.0181	25.939 0.148 - 2.104 0.830 1.084 - 11.504 - 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.622 0.075	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 0.0255 0.325 - - 0.946 0.154 0.495 0.154 0.495 0.171 0.255	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.102 0.3723 0.020 0.387 - - 0.992 0.102 0.282 - 0.282 - 0.183	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10, 0.40\\ \end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hf Nb Ta P Y La Ce Pr Nd Sm	25.001 0.151 2.150 0.475 0.125 0.013 13.356 0.0013 13.355 0.017 0.263 0.013 0.772 0.133 0.3066 0.047 0.182 0.0047	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.258 2.768 0.035 0.482 - 0.030 1.141 0.208 0.515 0.038 0.355 0.038	12.19 1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.290 0.290	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.265 0.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.031 0.344 0.031 0.346 0.269	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.096 - 13.558 0.021 3.708 0.031 0.254 - 0.029 0.804 0.147 0.404 0.147 0.404	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.074 0.025	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 0.055 0.325 - 0.946 0.154 0.495 0.171 0.255	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.200 0.387 - 0.992 0.102 0.282 - 0.183	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.161 3.015 0.047 0.451 	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\ -0.11\\ \end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hff Nb Ta P Y La Ce Pr Nd Sm Gd	25.001 0.151 2.150 0.475 0.125 0.013 13.356 0.0013 13.356 0.017 0.263 0.013 0.017 0.263 0.013 0.772 0.133 0.366 0.047 0.182 0.025	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.030 1.141 0.208 0.515 0.038 0.335 0.124	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.239 0.013	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.256 0.018 - 0.995 0.196 0.494 0.031 0.344 0.031 0.201	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.066	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.065 0.325 0.325 0.325 0.325 0.346 0.154 0.495 0.171 0.255	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 0.3723 0.202 0.387 - 0.992 0.102 0.282 - 0.183 - 0.093	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - 0.977 0.207 0.207 0.507 0.068 0.235	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\ -0.15\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hff Nb Ta P Y La Ce Pr Nd Sm Gd	25.001 0.151 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 0.013 0.772 0.133 0.366 0.047 0.182 0.200 0.502 0.200 0.502 0.201 0.255 0.255 0.127 0.125 0.125 0.125 0.125 0.017 0.125 0.125 0.017 0.125 0.125 0.017 0.125 0.125 0.017 0.125 0.125 0.017 0.125 0.125 0.017 0.255 0.017 0.256 0.013 0.013 0.256 0.027 0.133 0.366 0.047 0.182 0.020 0.020 0.027 0.132 0.266 0.027 0.132 0.266 0.027 0.132 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.267 0.272 0.272 0.272 0.272 0.272 0.277 0.267 0.277 0.267 0.277 0.267 0.277 0.267 0.277 0.267 0.277 0.267 0.277 0.277 0.267 0.2777 0.2777 0.2777 0.2777 0.2777 0.2777 0.2777 0.2777 0.2777 0.2777 0.27777 0.27777 0.27777 0.27777 0.277777 0.2777	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.038 2.768 0.030 1.141 0.208 0.515 0.038 0.335 0.110 0.122	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.091	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.031 0.344 0.080 0.081 0.005	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.251 3.708 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037 0.083 0.057	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.0075	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.022 0.427 0.013 - 1.074 0.225 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286 - 0.073 0.1073	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.0325 0.325 0.325 0.325 0.325 0.325 0.325 0.154 0.495 0.171 0.255 0.171	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102 0.282 - 0.183 - 0.0183 - 0.023	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 0.047 0.207 0.207 0.207 0.207 0.207 0.205 0.235 0.235	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.22-0.15\\ 0.$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hff Nb Ta P Y La Ce Pr Nd Sm Gd Dy Er	25.001 0.151 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 0.013 0.013 0.013 0.772 0.133 0.306 0.047 0.125 0.017 0.255 0.013 0.017 0.255 0.017 0.255 0.017 0.255 0.017 0.255 0.013 0.013 0.013 0.013 0.055 0.017 0.255 0.017 0.255 0.017 0.255 0.017 0.255 0.017 0.255 0.017 0.255 0.017 0.255 0.013 0.013 0.055 0.055 0.055 0.017 0.255 0.055	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.435 0.435 0.030 1.141 0.208 0.515 0.038 0.335 0.110 0.124 0.124 0.207	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.061 0.103 0.086 0.466 0.103 0.086 0.086	Hills +Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.196 0.494 0.031 0.344 0.081 0.025	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.025 13.708 0.025 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037 0.083 0.057 0.083	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.066 0.166	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.022 0.427 0.013 - 1.074 0.225 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286 - 0.073 0.107 0.012	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 0.325 0.325 0.325 0.325 0.346 0.154 0.495 0.171 0.255 0.171 0.255	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102 0.282 - 0.183 - 0.083 0.074 -	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 0.047 0.451 0.047 0.207 0.207 0.207 0.207 0.207 0.207 0.205 0.235	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.16\\ 0.03-0.16\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Na K Zr Hf Nb Ta P Y La Ce Pr Nd Sm Gd Dy Er	2 basis of (2 basis of (2.150 0.151 - 2.150 0.475 0.126 0.017 0.126 0.047 0.126 0.047 0.122 0.020 0.057 0.020 0.0772 0.020 0.0772 0.020 0.0777 0.0250 0.0777 0.0773 0.0050 0.0773 0.0775 0.0773 0.0775 0	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.035 0.482 - 0.030 1.141 0.2515 0.038 0.515 0.038 0.335 0.110 0.124 0.130	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.73 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.088 0.088 0.068	11110 +Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.260 3.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.031 0.344 0.081 0.025 0.114	= 29 24.761 0.247 - 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.477 0.404 0.075 0.181 0.037 0.083 0.057 0.054 0.054	25.939 0.148 - 2.104 0.830 1.084 0.247 2.339 - 0.547 0.027 0.029 1.095 0.1615 0.074 0.402 0.075 0.066 0.164 0.051	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.075 0.286 - 0.073 0.107 0.042 0.062	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 - - 0.946 0.154 0.495 0.171 0.255 - 0.063 0.145 0.059	24.754 0.099 0.017 1.568 0.435 0.067 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102 0.282 - 0.183 0.074 0.076 0.076	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - - 0.977 0.507 0.068 0.235 - - 0.026 0.120 0.071 0.071	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.11\\ 0.02-0.02\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hf Nb Ta F Y La Ce Pr Nd Sm Gd Dy Er Yb	2 basis of (2 basis of (2.150 0.151 - 2.150 0.475 0.125 0.013 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772 0.133 0.365 0.047 0.772 0.133 0.365 0.047 0.772 0.133 0.365 0.047 0.772 0.133 0.365 0.047 0.772 0.133 0.365 0.047 0.050 0.077 0.050 0.057 0.055 0.017 0.055 0.013 0.355 0.017 0.263 0.013 0.355 0.017 0.263 0.013 0.355 0.017 0.772 0.133 0.365 0.047 0.035 0.050 0.047 0.050 0.077 0.035 0.050 0.047 0.035 0.055 0.013 0.055 0.013 0.035 0.013 0.077 0.033 0.035 0.047 0.035 0.035 0.035 0.077 0.035 0.035 0.077 0.035 0.035 0.077 0.035 0.035 0.077 0.035 0.035 0.077 0.035 0.077 0.035 0.035 0.047 0.035 0.047 0.050 0.047 0.050 0.047 0.050 0.050 0.050 0.047 0.050 0.050 0.050 0.050 0.050 0.050 0.047 0.050 0.050 0.050 0.050 0.050 0.050 0.050 0.050 0.050 0.077 0.050 0.050 0.077 0.050 0.077 0.050 0.077 0.050 0.077 0.073 0.055 0.077 0.075 0.	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.035 0.482 - 0.030 1.141 0.208 0.515 0.035 0.482 - 0.030 1.141 0.208 0.335 0.110 0.124 0.130 0.086 0.704 0.158 0.158 0.158 0.251 0.158 0.251 0.158 0.251 0.158 0.258 0.258 0.258 0.258 0.258 0.158 0.258 0.258 0.258 0.258 0.258 0.355 0.482 - 0.030 0.1141 0.208 0.355 0.110 0.038 0.355 0.126 0.035 0.128 0.035 0.128 0.035 0.128 0.035 0.038 0.038 0.038 0.038 0.038 0.038 0.038 0.038 0.038 0.110 0.038 0.038 0.110 0.038 0.038 0.124 0.038 0.038 0.038 0.124 0.038 0.038 0.124 0.038 0.035 0.124 0.038 0.038 0.038 0.124 0.038 0.124 0.038 0.124 0.126 0.258 0.124 0.124 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.126 0.124 0.126 0.124 0.126 0.124 0.126 0.126 0.124 0.126 0.126 0.124 0.126 0.124 0.126 0.126 0.124 0.126 0.126 0.124 0.126	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.088 0.068 0.072		= 29 24.761 0.247 - 1.992 0.365 0.096 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.029 0.804 0.147 0.037 0.083 0.057 0.083 0.057 0.054 0.057 0.054 0.096	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.066 0.164 0.051 0.064	25.355 0.188 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.075 0.286 - 0.073 0.107 0.242 0.075 0.286	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 0.325 0.325 0.325 0.325 0.346 0.154 0.154 0.154 0.255 0.171 0.255 0.145 0.063 0.145 0.059 0.016	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102 0.992 0.102 0.282 - 0.183 0.074 0.074 0.076 0.021 1.568 0.074	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.11\\ 0.02-0.08\\ 1.56-1.96\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta F Y La Ce Pr Y La Ce Pr Nd Sm Gd Dy Er Yb Cl	2 basis of (2 5.001 0.151 - 2.150 0.475 0.125 0.125 0.125 0.125 0.125 0.13 13.356 0.305 3.555 0.017 0.263 0.013 - 0.772 0.133 0.366 0.047 0.772 0.133 0.366 0.047 0.772 0.133 0.366 0.047 0.772 0.133 0.366 0.047 0.772 0.133 0.366 0.047 0.772 0.153 0.355 0.013 0.355 0.017 0.772 0.153 0.366 0.047 0.151 0.151 0.151 0.155 0.125 0.017 0.263 0.013 0.356 0.007 0.772 0.133 0.366 0.047 0.151 0.151 0.155 0.017 0.772 0.133 0.366 0.047 0.153 0.355 0.017 0.772 0.133 0.365 0.047 0.133 0.366 0.047 0.133 0.366 0.047 0.151 0.133 0.366 0.047 0.133 0.366 0.047 0.133 0.366 0.047 0.133 0.366 0.047 0.133 0.366 0.047 0.133 0.366 0.047 0.132 0.055 0.077 0.155 0.026 0.035 0.077 0.163 0.055 0.077 0.163 0.055 0.077 0.073 0.055 0.077 0.073 0.035 0.035 0.035 0.047 0.073 0.035 0.035 0.035 0.047 0.073 0.035 0.035 0.047 0.077 0.073 0.035 0.046 0.047 0.162 0.077 0.073 0.035 0.045 0.045 0.045 0.045 0.047 0.055 0.045	Si+Ti+A 25.557 0.158	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.088 0.068 0.072 1.708		= 29 24.761 0.247 - 1.992 0.365 0.096 0.096 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037 0.083 0.057 0.054 0.019 1.781	25.939 0.148 - 2.104 0.830 1.084 0.247 2.339 - 0.547 0.027 0.029 1.095 0.1615 0.074 0.402 0.075 0.066 0.164 0.051 0.061 1.841	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.035 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286 - 0.073 0.107 0.042 0.050 1.639	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 - - 0.946 0.154 0.495 0.171 0.255 0.145 0.059 0.016 1.715	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.020 0.387 - 0.992 0.102 0.282 0.102 0.282 0.183 0.074 0.076 0.021 1.626	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - - 0.977 0.507 0.068 0.235 - - 0.026 0.120 0.071 0.030 1.786	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.68-1.35\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.11\\ 0.02-0.08\\ 1.56-1.86\end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na Sr Na K Zr Hf Nb Ta P Y La Ce Pr Y La Ce Pr Nd Sm Gd Dy Er Yb Cl	≥ basis of (25.001 0.151 2.150 0.475 0.125 0.013 13.356 0.0017 0.263 0.013 0.017 0.263 0.013 0.772 0.133 0.366 0.047 0.182 0.020 0.050 0.077 0.073 0.035 1.640 48.82	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.074 0.038 0.035 0.482 - 0.030 1.141 0.208 0.335 0.1124 0.130 0.086 0.070 1.704 48.50	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.088 0.068 0.072 1.708 -		= 29 24.761 0.247 1.992 0.365 0.096 - 13.558 0.031 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037 0.083 0.057 0.083 0.054 0.019 1.781 48.93	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.066 0.164 0.061 1.841 -	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.025 0.427 0.013 - 1.074 0.225 0.427 0.013 - 1.074 0.225 0.562 0.075 0.286 - 0.075 0.286 - 0.073 0.1074 0.225 0.422 0.026 0.716 0.222 0.2977 0.214 0.2981 0.2981 0.2981 0.2981 0.205 0.205 0.427 0.025 0.205 0.427 0.025 0.427 0.021 0.212 0.022 12.977 0.214 0.025 0.427 0.021 0.214 0.025 0.427 0.013 - 0.075 0.427 0.025 0.427 0.013 - 0.025 0.427 0.013 - 0.025 0.427 0.013 - 0.076 0.225 0.427 0.013 - 0.076 0.225 0.427 0.013 - 0.076 0.256 0.427 0.013 - 0.076 0.256 0.427 0.013 - 0.025 0.025 0.427 0.013 - 0.056 0.025 0.056 0.025 0.075 0.026 0.055 0.427 0.013 - 0.056 0.025 0.025 0.056 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.005 0.007 0.005 000	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 - - 0.946 0.154 0.495 0.171 0.255 0.171 0.255 0.171 0.255 0.063 0.145 5.0.53	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 3.723 0.200 0.387 - 0.992 0.102 0.282 - 0.183 0.783 0.76 0.083 0.076 0.021 1.626	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - 0.977 0.207 0.207 0.207 0.207 0.205 0.235 - 0.026 0.120 0.071 0.030 1.786 46.72	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.21-0.55\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.11\\ 0.02-0.08\\ 1.56-1.86\\ \hline \end{array}$
Cations on the Si Ti Al Fe Mn Ca Sr Na K Zr Hf Nb Ta P Y La Ce Pr Nd Sm Gd Dy Er Yb Cl Total ΣREE	2 basis of (2 5.001 0.151 2.150 0.475 0.125 0.013 13.356 0.0013 13.356 0.017 0.263 0.013 0.017 0.263 0.013 0.0772 0.133 0.366 0.047 0.182 0.020 0.050 0.077 0.073 0.035 1.640 48.82 0.98	Si+Ti+A 25.557 0.158 - 2.074 0.886 0.251 0.018 11.526 0.258 2.768 0.030 1.141 0.208 0.335 0.110 0.124 0.130 0.086 0.070 1.704 48.50 1.62	1+Zr+Hf- 25.171 0.111 - 1.618 0.480 0.173 - 12.940 0.297 3.211 - 0.463 0.044 - 1.257 0.166 0.448 0.029 0.239 0.091 0.103 0.088 0.068 0.072 1.708 - 48.78 - 1.31	+Nb+Ta) 25.204 0.269 - 2.368 0.710 0.505 - 14.469 0.225 0.028 0.256 0.018 - 0.995 0.196 0.494 0.081 0.081 0.081 0.114 0.041 1.787 51.50 1.41	= 29 24.761 0.247 1.992 0.365 0.096 - 13.558 0.251 3.708 0.031 0.254 - 0.029 0.804 0.147 0.404 0.075 0.181 0.037 0.083 0.054 0.019 1.781 48.93 1.06	25.939 0.148 - 2.104 0.830 1.084 - 11.504 0.247 2.339 - 0.547 0.027 0.029 1.095 0.189 0.615 0.074 0.402 0.075 0.066 0.164 0.051 0.061 1.841 49.43 1.70	25.355 0.188 - 2.209 0.716 0.222 0.022 12.977 0.214 2.981 0.025 0.427 0.013 - 1.074 0.225 0.427 0.013 - 1.074 0.225 0.427 0.075 0.286 - 0.073 0.1074 0.225 0.422 0.026 0.716 0.286 - 0.073 0.1074 0.225 0.286 - 0.073 0.1639 49.49 1.42	25.339 0.214 - 2.242 0.689 0.225 0.049 14.076 0.236 3.066 0.055 0.325 0.325 0.325 0.325 0.946 0.154 0.495 0.171 0.255 0.171 0.255 0.171 0.255 0.145 0.063 0.145 0.016 1.715 50.53 1.36	24.754 0.099 0.017 1.568 0.435 0.067 - 12.763 0.100 0.3723 0.020 0.387 - 0.992 0.102 0.282 - 0.183 - 0.083 0.076 0.021 1.626 47.37 0.32	25.332 0.137 0.018 2.047 0.454 0.070 0.023 10.927 0.168 3.015 0.047 0.451 - - 0.977 0.207 0.207 0.207 0.207 0.207 0.068 0.235 - 0.026 0.120 0.071 0.030 1.786 46.72 1.27	$\begin{array}{c} 24.67-25.94\\ 0.10-0.27\\0.02\\ 1.55-2.58\\ 0.36-0.94\\ 0.03-1.08\\0.05\\ 10.37-14.47\\ 0.10-0.35\\ 2.34-3.86\\0.06\\ 0.21-0.55\\0.04\\0.06\\ 0.21-0.55\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.23\\ 0.28-0.62\\0.17\\ 0.10-0.40\\0.11\\0.15\\ 0.03-0.16\\ 0.03-0.16\\ 0.03-0.11\\ 0.02-0.08\\ 1.56-1.86\\ \hline 46.7-51.5\\ 0.82-1.70\\ \end{array}$

 $FeO^* = total iron as Fe^{2+}$; bdl = below detection limit; N = number of analyses averaged.



Figure 15. Eudialyte-group mineral (EGM), BSE images from sample DV24828 Ailsa Craig microgranite. (a) A 20 µm EGM adjacent to fluorite (Fl) that filled an open space with quartz (Qz) enclosed by K-feldspar and albite (Kfs/Ab). (b) A 75 µm group of EGM filling a large open space in K-feldspar and albite (Kfs/Ab). The EGM appears coeval with quartz (Qz) and was followed by britholite (Bri). The sample preparation has exposed a vug (V). (c) A 22 µm EGM that has grown over a euhedral Zr-bearing aegirine (Aeg) enclosed by K-feldspar and albite (Kfs/Ab). (d) A 23 µm zoned EGM adjacent to a Zr-bearing aegirine (Aeg) enclosed by K-feldspar and albite (Kfs/Ab). The contrast in this image and in (e) has been adjusted to show the zoning. The bright rim has high REEs and lower Na₂O than the darker portion. (e) A 45 µm zoned EGM filling an open space enclosed by K-feldspar and albite (Kfs/Ab). The composition of the bright and dark portions is given in Table 13; the bright part has high REEs and lower Na₂O than the darker portion.

4.94 wt %–9.19 wt %. These values yield $\Sigma REE+Y$ that vary from 7.91 wt %–13.13 wt %, with Y dominant. Ailsa Craig microgranite EGMs, along with britholite-(Ce) (Macdonald et al., 2013b), monazite-(Ce), hydrothermal zircon, and a pyrochlore-group mineral, are the main REE carriers. Figure 16 shows the chondrite-normalized plot of the Ailsa Craig microgranite compared to EGMs. The microgranite is enriched slightly in LREEs with a flat HREE distribution, with [La/Yb]_{CN} in the range of 2.2–4.1. Although the REEs in EGM show variability, the overall patterns are similar to the bulk rock.



Figure 16. Chondrite-normalized plot comparing the REE pattern from Ailsa Craig granite AC-E (Govindaraju, 1987; Table 2) with measured range and average REEs from EGMs in this study (sample DV24828). Ho plotted in granite and Y plotted for Ho in EGMs. Normalizing factors from McDonough and Sun (1995).

EGMs are particularly prone to being partly to completely replaced by secondary phases during magmatic to hydrothermal stages (Mitchell and Liferovich, 2006; Sheard et al., 2012; Karup-Møller and Rose-Hansen, 2013; Wu et al., 2016; Estrade et al., 2014, 2018). In the Ailsa Craig microgranite, a positive correlation between analytical totals and Na₂O contents may indicate the slight loss of Na during secondary hydration. Such hydration might also explain low EPMA totals; otherwise, the EGMs seem not to have suffered significant replacement.

6.5 Zr-bearing fibrous mineral

Sheaflike bundles of Zr-bearing fibers occur in cracks and crevices in earlier formed minerals in the Ailsa Craig microgranite (Fig. 17). Individual fibers are generally $< 1 \,\mu$ m thick and a few micrometers long. The mode of occurrence suggests late-stage formation, and the texture indicates that it is not an alteration phase of its adjacent hosts.

The fiber surface was poorly polished and/or the fiber bundles were slightly recessed, making EPMA experiments questionable. Furthermore, the intrinsically high EPMA probe current produced some damage. We used EDS to attempt to determine composition. Concentrations were calculated from the count data using the factory standardization and the PeBaZAF quantification scheme recommended for samples with rough surfaces. The EDAX PeBaZAF correction scheme does not normalize to 100 %; thus it is helpful in identifying hydrous phases. The mean fiber composition is (N = 4) SiO₂ 52.8, TiO₂ 0.4, Al₂O₃ 5.3, FeO^{total} 5.6, MnO 0.8, MgO 0.4, CaO 4.8, Na₂O 4.3, K₂O 0.9, ZrO₂ 20.1, HfO₂ 2.4, and total 97.8 (wt %). All totals are low, suggesting a hydrous phase.



Figure 17. BSE images, Ailsa Craig microgranite, sample DV24828. (a) Typical texture of Zr-bearing fibrous phase (*ZrX*) in a vug in albite. (b) Zr-bearing fibers (*ZrX*), \sim 100–200 nm in width, in an open space between aegirine (Aeg) and K-feldspar (Kfs).

In the Ailsa Craig microgranite, Harding (1983) described fibrous glauconitic minerals of interstitial crystallization whose composition is typical for that mineral group with essentially nil ZrO₂. Piilonen et al. (1998) described fibrous aegirine, whose composition is depleted of ZrO₂ and TiO₂ and enriched in the aegirine component, unlike the Ailsa Craig Zr-bearing fibers. The literature description of the sheaflike habit of lemoynite ((Na,K)₂CaZr₂Si₁₀O₂₆ 5–6H₂O) (Perrault et al., 1969; Anthony et al., 1995) and other members of the lemoynite group (McDonald and Chao, 2001; McDonald et al., 2015) fits the morphology of the Ailsa Craig Zr-bearing fibers. Additionally, the SiO₂, ZrO₂, CaO, and Na₂O abundances of lemoynite are similar to those in the Ailsa Craig fibers. However, further work is needed to identify this Zrbearing fibrous silicate.

6.6 Zr-bearing aegirine

The Ailsa Craig microgranite has both late-stage Zr-bearing clinopyroxene and riebeckite–arfvedsonite. The Zr abundance in the amphibole is low, 0.62 wt % ZrO₂ (Harrison et al., 1987, their Table 2, mean of 48 analyses), and that concentration was confirmed with EDS. However, Harding (1983) reported that the clinopyroxene has up to 3.15 ZrO₂ wt %. Furthermore, the Zr-bearing clinopyroxene is frequently associated with EGM precipitation (Fig. 15c, d); thus, a reconnaissance study of the clinopyroxene was made. Many clinopyroxene grains were in cracks or poorly polished, limiting the EPMA study.

In Ailsa Craig sample DV24828, the clinopyroxene occurs as euhedral, subhedral, and interstitial grains; although texturally related to EGM, it is not associated with dalyite, at least not in our sample. Harding (1983) and Harrison et al. (1987) report a range of aegirine–hedenbergite compositions in various parts of the Ailsa Craig microgranite. Our analyses yielded aegirine compositions (Table 14). The compositions have a limited range and are intermediate between the low and high CaO clinopyroxenes reported by Harding (1983) in the microgranite. The T site shows almost complete occupancy by Si and varies from 1.98–1.99 apfu, and the Al content is very low. Low Al clinopyroxene, usually Zr-bearing, has been described from peralkaline rocks such as fenite (Ranløv and Dymek, 1981), peralkaline nepheline syenite (Shearer and Larsen, 1994; Preston et al., 1998; Murad, 2006), K-Na agpaitic syenite (Arzamastsev et al., 2000), peralkaline trachyte and comendite (Duggan, 1988), and trachyphonolite (Melluso et al., 2012). The M2 site has limited Na \leftrightarrow Ca substitution; Na varies from 0.91–0.96 apfu, and Ca varies from 0.10–0.13 apfu. The Fe^{3+}/Fe^{2+} ratio was adjusted to compensate for charge balance considerations related to Na \leftrightarrow Ca substitution. MgO was consistently below detection; thus the aegirine composition varies between Ae and Hd. The ZrO_2 content varies from 2.2 wt %-3.1 wt %, yielding apfu from 0.041–0.057. Jones and Peckett (1980) suggested that Zr is incorporated in the pyroxene structure only when no other Zr-bearing phase, such as eudialyte, is formed. In contrast, we find a common textural association between Zr-aegirine and a Zr-bearing EGM in the Ailsa Craig sample, indicating that they are both late-stage phases forming during the same time. The aegirine analyses showed scattered REE contents at or just above the detection limit (Table 14). A worthwhile study would be to analyze the Ailsa Craig Zr-aegirine with LA-ICP-MS for a complete picture of the REE content.

7 Discussion

7.1 Crystallization sequence

Judging from the examples shown in the BSE images in Figs. 2, 7, 8, 9, and 10 the order of crystallization in the granites, excepting Ailsa Craig, was baddeleyite \rightarrow zirconolite \rightarrow zircon \rightarrow hydrothermal zircon. Crystals of baddeleyite, zirconolite, and zircon are clearly magmatic; they are euhedral and enclosed in early-crystallizing phases, and some may show magmatic zoning. The late-magmatic zircons mainly fill spaces between crystals and vugs, although there is still some magmatic zoning. In the Ailsa Craig granite, all Zr-bearing phases, dalyite, EGM, zircon, Zr-aegirine, and probable lemoynite are late-stage with no obvious textures that would suggest a definitive sequence. Hydrothermal

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ladie 14.	Compositions	of Zr-bearing	aegirine

	Ailsa Craig granite						
Analysis	E7.2	E12.1	E12.2	E13.1			
N =	1	1	1	1			
SiO_2 (wt %)	52.06	52.22	51.83	52.31			
TiO ₂	1.51	1.68	0.84	0.87			
Al ₂ O ₃	0.21	0.26	0.20	0.19			
$Fe_2O_3^*$	28.13	28.16	28.55	28.83			
MnO	0.49	0.50	0.39	0.38			
MgO	bdl	bdl	bdl	bdl			
CaO	3.14	2.68	2.49	2.56			
SrO	bdl	bdl	bdl	0.07			
Na ₂ O	12.26	13.09	12.80	12.76			
$\tilde{K_2O}$	0.02	bdl	bdl	bdl			
ZrO_2	2.19	2.50	2.66	3.07			
HfO_2	0.20	0.21	0.16	0.29			
Nb_2O_5	0.22	0.27	bdl	bdl			
$Y_2 \tilde{O}_3$	0.12	bdl	bdl	bdl			
La_2O_3	bdl	bdl	bdl	0.10			
Pr_2O_3	bdl	bdl	0.14	bdl			
Nd_2O_3	0.23	bdl	bdl	0.31			
Sm_2O_3	bdl	0.16	bdl	bdl			
Gd_2O_3	bdl	bdl	0.24	bdl			
Yb_2O_3	bdl	bdl	bdl	0.14			
Total	100.78	101.73	100.30	101.89			
Formulae on t	he basis of	f six oxyge	ens and fou	r cations			
Si	1.991	1.981	1.996	1.990			
Ti	0.044	0.048	0.024	0.025			
Al	0.009	0.011	0.009	0.009			
Fe ³⁺	0.732	0.721	0.756	0.762			
Fe ²⁺	0.113	0.093	0.090	0.092			
Mn	0.016	0.016	0.013	0.012			
Mg	_	_	_	_			
Ca	0.129	0.109	0.103	0.104			
Sr	_	_	_	0.002			
Na	0.909	0.963	0.956	0.941			
Κ	0.001	_	_	_			
Zr	0.041	0.046	0.050	0.057			
Hf	0.002	0.002	0.002	0.003			
Nb	0.004	0.005	_	_			
Y	0.002	_	_	_			
La	_	_	_	0.001			
Pr	_	_	0.002	_			
Nd	0.003	_	_	0.004			
Sm	_	0.002	-	-			
Gd	_	_	0.003	_			
Yb	_	_	0.000	0.002			
Σcations	4.00	4.00	4.00	4.00			
07 A **	87	89	89	89			

 $Fe_2O_3^* = all Fe as Fe^{3+}; bdl = below detection limit. Fe^{3+}/Fe^{2+} based on stoichiometry. <math>N =$ number of analyses averaged. Ta₂O₅, Ce₂O₃, Dy₂O₃, Er₂O₃ = bdl. ** = method of Morimoto (1989).

zircon is present in several forms, and a major feature is the high concentration of incompatible elements and the abundance of inclusions. We observed no texture that suggested that early formed magmatic zircon was altered by late-stage solutions to yield either hydrothermal type 1 or type 2.

7.2 Mineral parageneses

Zirconolite and baddeleyite occur in a wide range of terrestrial rocks, including carbonatites, kimberlites, mafic and ultramafic intrusions, dolerite dikes, alkaline intrusions, pegmatites, potassic lavas, skarns, metacarbonates, and granulites. The textural relationships between them and with zircon are highly variable. Barkov et al. (1995) found zircon replacing baddeleyite in the Imandrovsky mafic intrusion, Kola Peninsula, Russia. Baddeleyite and zirconolite have been reported as coexisting, although not necessarily in equilibrium, in, inter alia, intercumulus areas in the Rhum (Rum) layered intrusion, Scotland (Williams, 1978). Williams (1996) showed that zirconolite crystallized later than, and sometimes replaced, baddeleyite in phoscorite of the Kovdor carbonatite complex, Kola Peninsula, Russia. In potassic rocks of the Lewotolo volcano, Indonesia, the three phases apparently crystallized in the sequence zircon \rightarrow zirconolite \rightarrow baddeleyite (de Hoog and van Bergen, 2000). Most sills and dikes emplaced during the Franklin episode in northern Canada contain zircon, baddeleyite, and zirconolite; some zirconolite grains apparently precipitated with, or later than, baddeleyite (Heaman et al., 1992). In carbonatite veins of the Jacupiranga complex, southern Brazil, baddelevite occurs as tabular crystals containing zirconolite (Salvioli-Mariani et al., 2012). Zhao et al. (2016) reported baddeleyite and zircon precipitated from hydrothermal fluids during skarnization at the Tengtie iron deposit, South China.

In the UK granites of this study, baddeleyite and zirconolite crystallized in the type 1 (biotite \pm amphibole) and type 2 (hedenbergite–fayalitic olivine) granites, and the order of crystallization, as noted above, was baddeleyite \rightarrow zirconolite \rightarrow zircon. The textural relationships of the zirconolite (also baddeleyite) in the Ardnamurchan samples are similar to those described for St. Kilda and lunar zirconolite, i.e., precipitation from late-stage liquids in which HREE concentration is high and silica activity is low enough to form Zr oxides and not zircon.

Various factors have been invoked to explain the stability relationships of the three phases:

Silica activity. In reversible, anhydrous, relatively high-temperature experiments, the equilibrium reaction is ZrO₂+SiO₂ ↔ ZrSiO₄. Yet, in both the Rum and Ard-namurchan samples, baddeleyite is commonly hosted by quartz (e.g., Fig. 10a), and at high magnification using FE-SEM, no incipient reaction on the baddeleyite–quartz interface was observed. While aSiO₂ in the UK granites may have varied from the adamellites to the alkali feldspar granites, the ubiquitous presence of quartz

may have meant that the activity was always close to unity. Perhaps the crystallization of baddeleyite was a result of local metastability.

ii. Calcium activity. In a study of zircon-bearing veins in a harzburgite xenolith from a kimberlite, Dawson et al. (2001) proposed that zircon was an early precipitate from the melt but, after a desilication reaction, was replaced by baddeleyite. The magma was then Ca-Ti metasomatized, and zirconolite formed. Carlier and Lorand (2008) noted in lamprophyric melts of the south Peruvian potassic province the crystallization sequence Zr-rich titanite \rightarrow perrierite \rightarrow zirconolite + baddelevite and ascribed the changes to decreasing Ca activity. In a series of experiments, Lewerentz et al. (2019) reacted unaltered zircon with H2O-CaCl2 and H2O-Ca(OH)₂ fluids. They found that baddeleyite replaced the zircon only when Ca saturated the system, lowering the $aSiO_2$ to < 1. However, they noted that in natural rocks, baddeleyite can form in the interiors of zircon in contact with quartz, which they attributed to local, small-scale metastability. The average CaO abundances in the granites are type 1 = 1.78 wt %; type 2 =1.23 wt %; and type 3 = 0.43 wt %. If the concentrations can be taken as reasonable estimates of Ca activity, this could explain the presence of zirconolite in types 1 and 2 but its absence in type 3, peralkaline granites.

Dalvite has been recorded as an accessory mineral in a range of rocks, including peralkaline granites and syenites and highly potassic rocks such as lamproites and lamprophyres. Harris et al. (1982) reported dalyite in a peralkaline granite block from Ascension Island. In the Strange Lake peralkaline granite complex, Quebec-Labrador, dalyite nucleated heterogeneously on older zircon crystals (Salvi and Williams-Jones, 1995). Jeffery et al. (2016) reported dalyite in syenitic ejecta of the Caldeira-Castelinho Ignimbrite Formation of Terceira, Azores, where it is interstitial, (partly) filling void spaces, as shown by some of the Ailsa Craig crystals (Fig. 13). Dalyite crystallization is thus favored in peralkaline silica-oversaturated conditions, as in the Ailsa Craig sample, and probably in melts with a high chemical potential of K₂O (µK₂O; Marks et al., 2011; Jeffery et al., 2016). In the Ailsa Craig sample studied, there is no obvious crystallization sequence or transition among eudialyte, dalyite, or zircon based on texture; all appear as late-stage.

Eudialyte-group minerals are recorded most frequently in strongly silica-undersaturated alkaline rocks, such as agpaitic nepheline syenites and urtites. However, it is also known from alkaline granites and associated pegmatites; Estrade et al. (2018) list eight occurrences worldwide. Eudialyte was reported, for example, in a granite block from Ascension Island, where it occurs adjacent to dalyite with no evidence of reaction (Harris et al., 1982). Like dalyite, an unspecified eudialyte-group mineral totally or partially infills interstitial pore spaces in the Terceira syenites, but the two are not seen in contact (Jeffery et al., 2016). Among the UK Paleogene granites, eudialyte sensu lato was recorded in quartz syenites of the Carlingford complex, Northern Ireland, by Nockolds (1950), occurring in segregation areas with quartz, aegirine augite, and titanite. It was also found in crushes of the Rock-all peralkaline granite (Sabine, 1960).

According to Sørensen (1997), the main factors controlling the crystallization of eudialyte-group minerals in a melt are its peralkalinity and high Zr and volatile contents, mainly F, Cl, and H₂O. Noting the high Cl abundances in eudialytegroup minerals, Estrade et al. (2018) specified that high Cl contents in the melt are a critical parameter for the crystallization of eudialyte in peralkaline granites.

8 Summary

This study documents the zirconium mineral sequence in UK Paleogene granites, types 1, 2, and 3, from early magmatic (baddeleyite \rightarrow zirconolite \rightarrow zircon) through latemagmatic to hydrothermal zircon. In the Ailsa Craig granite, all the Zr-bearing phases, dalyite, eudialyte, hydrothermal zircon, probable lemoynite, and Zr-aegirine, are late-stage phases with no textural evidence for a definitive sequence.

The more strongly peralkaline Ailsa Craig microgranite contains a REE+Y-rich, Ca-poor eudialyte-group mineral and dalyite, the first documented occurrence in the United Kingdom. A late-stage incompatible-element-enriched hydrous zircon and a Zr-bearing fibrous mineral, probable lemoynite, were identified.

Zirconolite in the Grigadale Granophyre and Granophyre Centre 2 from Ardnamurchan and the Glamaig Granite from Skye Western Red Hills Centre are exceptionally REE+Yrich with Y dominant.

Data availability. All data are provided in the tables and in the Supplement.

Supplement. The supplement related to this article is available online at: https://doi.org/10.5194/ejm-33-537-2021-supplement.

Author contributions. RM obtained the samples. HEB examined and described petrographically the samples with SEM and analyzed the phases with electron microprobe. Both RM and HEB contributed to the text.

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