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# The unique Katugin rare-metal deposit (southern Siberia): Constraints on age and genesis



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#### ABSTRACT

We report new geological, mineralogical, geochemical and geochronological data about the Katugin Ta-Nb-Y-Zr (REE) deposit, which is located in the Kalar Ridge of Eastern Siberia (the southern part of the Siberian Craton). All these data support a magmatic origin of the Katugin rare-metal deposit rather than the previously proposed metasomatic fault-related origin. Our research has proved the genetic relation between ores of the Katugin deposit and granites of the Katugin complex. We have studied granites of the eastern segment of the Eastern Katugin massif, including arfvedsonite, aegirine-arfvedsonite and aegirine granites. These granites belong to the peralkaline type. They are characterized by high alkali content (up to  $11.8 \text{ wt}\% \text{ Na}_2\text{O} + \text{K}_2\text{O}$ ), extremely high iron content (FeO\* + MgO) = 0.96-1.00), very high content of most incompatible elements - Rb, Y, Zr, Hf, Ta, Nb, Th, U, REEs (except for Eu) and F, and low concentrations of CaO, MgO, P<sub>2</sub>O<sub>5</sub>, Ba, and Sr. They demonstrate negative and CHUR-close  $\varepsilon$ Nd(t) values of 0.0... – 1.9. We suggest that basaltic magmas of OIB type (possibly with some the crustal contamination) represent a dominant part of the granitic source. Moreover, the fluorine-enriched fluid phases could provide an additional source of the fluorine. We conclude that most of the mineralization of the Katugin ore deposit occurred during the magmatic stage of the alkaline granitic source melt. The results of detailed mineralogical studies suggest three major types of ores in the Katugin deposit: Zr mineralization, Ta-Nb-REE mineralization and aluminum fluoride mineralization. Most of the ore minerals crystallized from the silicate melt during the magmatic stage. The accessory cryolites in granites crystallized from the magmatic silicate melt enriched in fluorine. However, cryolites in large veins and lens-like bodies crystallized in the latest stage from the fluorine enriched melt. The zircons from the ores in the aegirine-arfvedsonite granite have been dated at 2055  $\pm$  7 Ma. This age is close to the previously published 2066  $\pm$  6 Ma zircon age of the aegirine-arfvedsonite granites, suggesting that the formation of the Katugin rare-metal deposit is genetically related to the formation of peralkaline granites. We conclude that Katugin rare-metal granites are anorogenic. They can be related to a Paleoproterozoic ( $\sim 2.05$  Ga) mantle plume. As there is no evidence of the 2.05 Ga mantle plume in other areas of southern Siberia, we suggest that the Katugin mineralization occurred on the distant allochtonous terrane, which has been accreted to Siberian Craton later.

#### 1. Introduction

The Katugin Ta-Nb-Y-Zr (REE) deposit is located in the Kalar Ridge

of Eastern Siberia, Russia (Fig. 1). This is one of the largest Precambrian Ta-Nb-Y deposits in the world and it belongs to the category of unique deposits containing industrial concentrations of  $ZrO_2$  (1.51%),  $Nb_2O_5$ 

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Fig. 1. The major tectonic elements of the Siberian craton, modified after Gladkochub et al. (2006) and Rosen (2003) (a); schematic geological map of the Udokan subzone of the Kodar–Udokan trough, modified after Podkovyrov et al., 2006 (b).

(0.30%), Ta<sub>2</sub>O<sub>5</sub> (0.026%), Y<sub>2</sub>O<sub>3</sub> (0.16%), U and REE (Arkhangel'skaya et al., 1998). One segment in the southern part of the deposit is extremely enriched in Ta<sub>2</sub>O<sub>5</sub> (> 1.0% in average) (Pirajno and Santosh, 2014).

The Katugin deposit was discovered in the middle of the 20th century and has been explored over several decades. As most of the deposit is poorly exposed, tens of exploration holes have been drilled, two exploration tunnels and tens of trenches have been excavated. Prospecting and an economic appraisal of the Katugin deposit were completed in 1990s, with most of drill cores having been stored and used for new research. Any geological information about the Katugin deposit was hardly available to the international community until recently - most having been published in "grey" literature (e.g. internal reports). The only valuable exception was the paper of Arkhangel'skaya et al. (1993), but it was detailed enough for a comprehensive understanding of the genesis and specific aspects of the deposit. From the very beginning of its exploration the Katugin deposit was considered to belong to a metasomatic type controlled by faults (Prokhorov and Sobachenko, 1985; Arkhangel'skaya et al., 1993). Meanwhile Larin et al. (2002) identified the ore-bearing and ore-controlling rocks as the A-type alkaline granites and determined their Paleoproterozoic age. The characteristics of these granites and ores have been analysed and reported in a series of recent publications (Levashova et al., 2014; Kotov et al., 2015a,b; Larin et al., 2015; Sklyarov et al., 2016). These workers argue in favour of a genetic relation between Nb-Ta-Zr-Y (REE) ores of the Katugin deposit and alkaline granites of the Katugin complex, as a counter to a previously published hypothesis of their metasomatic fault-related genesis.

Here we systematize those arguments and present new data about types and age of mineralization, and propose our hypothesis of the genesis of the unique Katugin Ta-Nb-Y-Zr (REE) deposit.

#### 2. Geological background

The Siberian craton is richly endowed with rare-metal mineral deposits (see overviews of Seltmann et al., 2010; Pirajno and Santosh, 2014). The most famous – Tomtor, Zaschikhinskoe and Beloziminskoe deposits are related to carbonatite. Whereas others, such as Vishnia-kovskoe and Goltsovoe are considered to be related to rare-metal granite and pegmatite. However, these issues for the Katugin deposit are still under discussion.

The Katugin deposit is located in the northern margin of the Stanovoy suture zone which separates the Aldan Shield (southern part of the Siberian craton) and adjacent terranes of the Central Asian Orogenic Belt (Fig. 1a). The mineralization is localized in two rather small (3 and 18 km<sup>2</sup>) areas (Fig. 1b), which were previously considered either as fault-related metasomatic zones (Arkhangel'skaya et al., 1993, 2012; Bykov and Arkhangel'skaya, 1995; Prokhorov and Sobachenko, 1985; Ryabtsev et al., 2006), or as a massifs of Paleoproterozoic alkaline granites of the Katugin complex (Larin et al., 2002, 2015; Levashova et al., 2014).

The eastern part of the massif (hereafter Eastern Katugin massif) is better exposed than its western part. Moreover, hundreds of bore-holes were drilled here and two adits were mined. In this study we present new data from the Eastern Katugin massif only.



Fig. 2. Schematic geological map of the Eastern Katugin massif, modified after Prokhorov and Sobachenko (1985).

The Eastern Katugin massif has a curved heart-like shape and consists of two segments (Fig. 2). We have to note that the published maps (Arkhangel'skaya et al., 1993, 2012 etc), including the one used for our Fig. 2, are partly motivated by the metasomatic model of the massif's genesis and suffer from several contradictions. First, the metasomatite shape is overlain by granite and blastomylonite configurations without showing outlines of the bodies. Second, jagged contacts between different kinds of rocks hardly reflect the real situation, but mainly caused by difficulties in the correlation between boreholes and extensive complexity of the lithological varieties. Nonetheless, these maps are still useful for understanding the geometry and composition of the massif and related positions of ore bodies.

The eastern segment of the Eastern Katugin massif is composed by aegirine (Aeg), aegirine-arfvedsonite (Aeg-Arf), arfvedsonite (Arf) granites and to a less extent by biotite-arfvedsonite (Bt-Arf) granites. Biotite (Bt) and riebeckite-biotite (Rbk-Bt) granites compose the western segment, with biotite being dominant in the melanocratic rocks. Only the narrow contact zone in the northern part of this segment is composed by Arf and Aeg-Arf granites, and it is possible that this narrow strip is the continuation of the eastern segment.

Most of granites in the Eastern Katugin massif are medium- to finegrained, but coarse-grained and even pegmatite-like varieties are also found in forms of cutting veins of irregular shapes with thicknesses of between 0.15 and 3 m. A foliation fabric of different extent is characteristic to all types of granites. Sometimes the foliation is visible only in a macroscopic scale (rocks look like normal massive granites under microscope), but more often the foliation is expressed in the microscopic level as well by lineation of biotite or arfvedsonite (aegirine) grains. Probably this widespread foliation was the reason for the initial hypothesis of the metasomatic genesis of the intrusive rocks. The occurrence of fragments of the host biotite gneisses within granites and their similar mineralogical composition probably also provoked a "metasomatic" hypothesis. A petrographic study of the foliated granites demonstrated that in most cases their foliation resulted during emplacement, not during a metamorphic reworking. Dark minerals do not have any traces of re-crystallization and their composition is the same both in massive and in foliated rocks.

Fig. 2 clearly demonstrates that zonation from aegirine through amphibole to biotite rocks, suggested in earlier studies (Arkhangel'skaya et al., 1993, 2012), is only apparent. In fact the eastern segments of the intrusive massif, including the narrow northwestern strip, are mostly composed by Aeg, Aeg-Arf and Arf granites, whilst biotite and Rbk-Bt granites are dominant in the western segment. This could be explained by two pulses of granite emplacement.

#### 3. Methods

> 120 samples of granites from the eastern segment of the Eastern Katugin massif have been collected to study their mineral and geochemical composition. Nine samples were taken to analyse the isotopic compositions of Nd, and to identify possible rock sources and to calculate their Nd-model age. We also made U-Pb zircon age determination of ore-segregations located among granites of the Eastern Katugin massif.

For the preliminary identification of minerals we analysed energydispersive spectra (EDS), back-scattered electron (BSE) images and elemental mapping (EDS system), using a TESCAN MIRA 3MLU scanning electron microscope equipped with a microprobe system INCA Energy 450 XMax-80 (Oxford Instruments Ltd) at the Institute of Geology and Mineralogy of the Siberian Branch of Russian Academy of Sciences (IGM SB RAS), Novosibirsk, Russia. Analytical parameters: high-vacuum mode, an accelerating voltage of 20 kV, a probe current of 1 nA, counting time of 20–60 s. Electron microprobe analyses (EMPA-WDS) of minerals have been carried out in IGM SB RAS, using a JEOL JXA-8100 electron microprobe. The operating conditions were as follows: beam diameter of  $1-2 \mu m$ , accelerating voltage of 20 kV and a beam current of 10–15 nA, counting time of 10 s (for all elements). Precision for the major elements was better than 2%. For the data reduction we used the PAP routine (Pouchou and Pichoir, 1985).

We studied the content of Be, Li, B, Rb, Cs, Ba, Sr, Zr, Nb, Sn and H<sub>2</sub>O in, amphibole, aegirine and astrophyllite of the Katugin deposit using the secondary-ion mass spectrometry (SIMS) on the Cameca IMS-4f ion probe in the Analytical Centre of the Yaroslavl Branch of the Institute of Physics and Technology (Yaroslavl, Russia). Only the previously analysed by EMPA  $> 20 \,\mu m$  grains have been used. Analysis of trace elements was carried out by the energy filter method; the operating conditions were as follows: primary  $O^{2-}$  beam – 20  $\mu$ m, I = 2-4 nA, energy offset - 100 eV, and energy slit - 50 eV. Concentrations of elements were determined from the ratios of their isotopes to <sup>30</sup>Si, using calibration curves for standard samples (Jochum et al., 2000). The content of hydrogen has been determined from the <sup>1</sup>H mass together with trace elements. Low background content of H<sub>2</sub>O (0.03 wt%) in the mass spectrometer was achieved by storing the samples for 24 h in high-vacuum. The NIST610 glass was used as an external standard.

Major elements and F in granites of the Eastern Katugin massif were determined by wet chemistry at the Centre for Geodynamics and Geochronology of the Institute of the Earth's Crust (Irkutsk).  $Al_2O_3$  were analysed by atomic absorption spectroscopy (AAS) in samples of 14231, 14232, 14233, 14234, 14235, 14240, 14241, 14248, C-94-1, C-265-1, C-147-1, C-54-36 at the Vinogradov Institute of Geochemistry (Irkutsk). Trace elements and rare earths were determined by inductively coupled plasma mass spectrometry (ICP-MS) on an Agilent 7500ce (USA) at the Limnological Institute (Irkutsk). For ICP-MS the samples were fused with LiBO<sub>2</sub> following the procedure of Panteeva et al. (2003). Calibrations were with internal and international standards G-2, GSP-2, JG-2, and RGM-1. Analytical accuracy was 0.5–1.0% for major oxides and up to 5% for trace elements and REE.

Sm and Nd were extracted using the technique described by Kotov et al. (1995). Isotopic measurements were carried out at the IPGG RAS

(St. Petersburg, Russia) using multicollector Triton T1 and Finnigan MAT 261 mass spectrometers. The measured  $^{143}\rm Nd/^{144}\rm Nd$  values were normalized to  $^{146}\rm Nd/^{144}\rm Nd = 0.7219$  and reduced to  $^{146}\rm Nd/^{144}\rm Nd = 0.511860$  in the La Jolla Nd standard. The accuracy of the Sm and Nd measurements was 0.5%, and that of the  $^{147}\rm Sm/^{144}\rm Nd$  and  $^{143}\rm Nd/^{144}\rm Nd$  values was 0.005% (2 $\sigma$ ). The average weighted  $^{143}\rm Nd/^{144}\rm Nd$  value in the JNdi Nd standard derived from seven measurements is 0.512104  $\pm$  8. The blank run was 0.03–0.2 and 0.1–0.5 ng for Sm and Nd, respectively. When calculating the values of the  $\epsilon_{\rm Nd}(t)$  and model  $t_{\rm Nd}(\rm DM)$  ages, we used the recent values accepted for the uniform chondrite reservoir CHUR (Jacobsen and Wasserburg, 1984) and depleted mantle DM (Goldstein and Jacobsen, 1988). For calculating two-stage model ages  $t_{\rm Nd}(\rm DM-2st)$ , the average crust value of  $^{147}\rm Sm/^{144}\rm Nd = 0.12$  has been used.

Zircon separation was carried out following the standard procedures applying heavy liquids. Its morphological features were examined with the optic microscope (Leica DMPL) at 500-1000 times magnification ant with the ABT 55 and VEGA3 TES SCAN scanning electron microscopes employing secondary electron and cathodoluminescence detectors. The zircon crystals selected for U-Pb analysis were subjected to multistep removal of surficial contaminants using alcohol, acetone, and 1 M HNO<sub>3</sub>. Between each step the zircons and their fragments were rinsed in ultrapure water. Chemical dissolution zircons and isolation of U and Pb were carried out following the modified method of Krogh (1973). In some cases in order to diminish the degree of discordance, an air-abrasive technique (Krogh, 1982) was applied. Isotope analyses have been performed using the multicollector Finnigan MAT 261 and Triton T1 mass spectrometers in both the static and the dynamic modes. The precision of determination of the concentrations of U and Pb as well as that of the Pb/U values was 0.5%. Blank contamination did not exceed 10 pg for Pb and 1 pg for U. The acquired raw data were processed with the PbDAT (Ludwig, 1991) and ISOPLOT software (Ludwig, 1999). For age calculation conventional U decay constants (Stacey and Kramers, 1975) were used. The correction for nonradiogenic lead was carried out in accordance with the model values (Steiger and Jager, 1976). All the errors presented are  $2\sigma$ .

### 4. Petrography and mineralogy of the Katugin granites from the eastern segment of the Eastern Katugin massif

In the Katugin complex from the eastern segment of the Eastern Katugin massif the Bt-Arf, Arf, Aeg-Arf and Aeg granites are distinguished. The drill core study demonstrated that the Bt-Arf granites form isolated zones in the massif (see Fig. 2). The Aeg, Aeg-Arf and Arf granites interchange with each other, exhibiting gradual transition zones between purely aegirine and purely arfvedsonite granites, and the aegirine/arfvedsonite ratio sometimes changes significantly within short distances of 3–5 m. No Bt-Arf granites have been found in such zones. Despite the Bt-Arf granites being enriched in HFSE and REE, they cannot be considered as the ore-bearing rocks of the Katugin rare-metal deposit, so we do not discuss them any more in this study.

The studied Arf, Aeg-Arf and Aeg granites are mainly composed of quartz (20–35%), albite (20–30%) and K-feldspar (38–48%), in addition to mafic minerals (aegirine and arfvedsonite). All rock-forming, minor and accessory minerals of the Katugin granites are listed in Table 1. The Arf granites contain arfvedsonite (1–8%) as the only dark-colored mineral. Biotite is absent, rare small aegirite grains are found. The Aeg-Arf granites contain aegirine (1–8%) and arfvedsonite (1–8%) in various proportions – sometimes one of these minerals can be dominant, in other cases their content is equal. The content of aegirine in the purely Aeg granites, this mineral is clearly later than arfvedsonite and aegirine. Minor astrophyllite, bafertisite and polylithonite are locally recorded. Fe-chlorite is the major secondary mineral.

High concentrations of zircon and pyrochlore, as well as the presence of cryolite, are characteristic for the Aeg and Aeg-Arf granites.

#### Table 1

Minerals in Arf, Aeg-Arf and Aeg granites from the eastern segment of the Eastern Katugin massif, cryolitic rocks and as crystalline inclusions in zircons, modified after Sklyarov et al. (2016).

Mineral	Formula	Granites	Inclusions in Zrn	Cryolitic rocks
Quartz	SiO <sub>2</sub>	+	+	+
K-feldspar	KAlSi <sub>3</sub> O <sub>8</sub>	+	+	+
Albite	NaAlSi <sub>3</sub> O <sub>8</sub>	+	+	+
Zircon	ZrSiO <sub>4</sub>	+	-	+
Thorite	ThSiO <sub>4</sub>	+	+	+
Polylithionite	KLi <sub>2</sub> AlSi <sub>4</sub> O <sub>10</sub> (F,OH) <sub>2</sub>	+	-	+
Hydrotetraferriannite	$(H_2O,K)Fe_3FeSi_3O_{10}(OH)_2$	+	-	-
Aegirine	NaFeSi <sub>2</sub> O <sub>6</sub>	+	+	+
Na-Fe-amphibole (group)	(Na,K)Na <sub>2</sub> Fe <sub>4</sub> <sup>2+</sup> Fe <sup>3+</sup> Si <sub>8</sub> O <sub>22</sub> (F,OH) <sub>2</sub>	+	-	+
Astrophyllite	(K,Na) <sub>3</sub> (Fe,Mn,Zn) <sub>7</sub> (Ti,Nb,Sn) <sub>2</sub> Si <sub>8</sub> O <sub>24</sub> (O,OH) <sub>7</sub>	+	-	+
Bafertisite	$Ba_2(Ti,Sn)_2(Fe,Mn)_4(Si_2O_7)_2O_2(OH)_2(F,OH)_2$	+	-	+
Fe-chlorite	$(Fe^{2+},Fe^{3+})_5Al(Si_3Al)O_{10}(OH,O)_8$	+	+	-
Pyrochlore (group)	(Ca,Na,REE,Y,Pb) <sub>2</sub> Nb <sub>2</sub> O <sub>6</sub> (OH,F)	+	+	+
Columbite-(Fe-Mn)	(Fe,Mn)Nb <sub>2</sub> O <sub>6</sub>	+	-	+
Fergusonite-(Y)	YNbO <sub>4</sub>	+	+	-
Samarskite-(Y)	(Y,Ce,U,Fe) <sub>3</sub> (Nb,Ta,Ti) <sub>5</sub> O <sub>16</sub>	+	+	-
Euxenite-(Y)	(Y,Ca,Ce,U,Th)(Nb,Ta,Ti) <sub>2</sub> O <sub>6</sub>	+	-	-
Ferberite	(Fe,Mn)WO <sub>4</sub>	+	-	+
Pseudobrookite	Fe <sub>2</sub> TiO <sub>5</sub>	+	-	-
Pseudorutile	Fe <sub>2</sub> Ti <sub>3</sub> O <sub>9</sub>	+	-	_
Ilmenite	(Fe,Mn)TiO <sub>3</sub>	+	-	+
Magnetite	FeFe <sub>2</sub> O <sub>4</sub>	+	-	+
Rutile	110 <sub>2</sub>	+	-	+
Cassiterite	SnO <sub>2</sub>	+	+	+
Certanite-(Ce)	$(Ce,Th)O_2$	-	-	+
Goethite	(Fe,Mn)O(OH)	+	+	+
Siderite	(Fe,MII)CO <sub>3</sub>	+	+	+
Rhodochrosite	(Mn,Fe)CO <sub>3</sub>	-	—	+
Calcite Restruccite (Co)	CaLO <sub>3</sub>	+	_	+
Barisite (Ce)	$(Ce, La, Nd)(CO_3)F$	+	+	+
Synchysite (Ce)	$C_2(C_2,L_3,Nd)_2(CO_3)_{3\Gamma_2}$	т _		
Fluorapatite	$C_2$ (BQ) F	_ _	- -	
Monazite (Ca)	$Ca_5(rO_4)_{31}$	+ +		
Xenotime_(V)	VPO.	+	+	_
Xenotime-(Vb)	(Yh Y)PO	_	+	_
Cheralite-(Ce)	$(Ce Th Ca)(P Si)O_4$	+	_	-
Barite	(Ba.Sr)SO4	+	_	+
Galena	PbS	+	+	_
Sphalerite	ZnS	+	_	+
Molibdenite	MoS <sub>2</sub>	+	_	_
Pyrrotite	Fe <sub>1-v</sub> S	+	_	_
Pvrite	FeS <sub>2</sub>	+	_	+
Chalcopyrite	CuFeS <sub>2</sub>	+	_	+
Bornite	$Cu_5FeS_4$	+	_	_
Bithmutine	Bi <sub>2</sub> S <sub>3</sub>	+	+	-
Lead	Pb	+	_	-
Bismuth	Bi	+	+	-
Fluorite + Y-fluorite	(Ca,Y)F <sub>2</sub>	+	+	+
Cryolite	Na <sub>3</sub> AlF <sub>6</sub>	+	+	+
Elpasolite	K <sub>2</sub> NaAlF <sub>6</sub>	-	_	+
Simmonsite	Na <sub>2</sub> LiAlF <sub>6</sub>	-	_	+
Tveitite-(Y)	$Ca_{1-x}Y_{x}F_{2+x}, x \approx 0.3$	+	+	+
«Ba-Sr-Tveitite-(Y)»	$(Ba,Sr,Ca)_{1-x}Y_{x}F_{2+x}, x \approx 0.3$	-	_	+
Gagarinite-(Y)	NaCaYF <sub>6</sub>	+	+	+
Fluocherite-(Ce)	(Ce,La,Nd)F <sub>3</sub>	+	+	+
Chiolite	Na <sub>5</sub> Al <sub>3</sub> F <sub>14</sub>	-	+	+
Neighborite	NaMgF <sub>3</sub>	-	-	+
Weberite	Na <sub>2</sub> MgAlF <sub>7</sub>	-	-	+
Usovite	Ba <sub>2</sub> CaMgAl <sub>2</sub> F <sub>14</sub>	-	-	+
Ba- fluoraluminates	BaAlF <sub>4</sub> (OH), BaCa <sub>2</sub> AlF <sub>9</sub> , BaCa <sub>4</sub> AlF <sub>13</sub>	-	-	+
H <sub>2</sub> O-OH-fluoraluminates	Prosopite, thomsenolite, pachnolite etc.	+	-	+

Note: +, mineral is present; -, mineral is absent.

Cryolite mostly occurs as streaks and disseminations, or as clusters in the Arf, Aeg-Arf and Aeg granites, but the large isolated cryolite lensshaped 10 m thick body is found in the eastern segment of the Eastern Katugin massif. This body has been traced through 200 m along strike.

The dark-colored minerals have similar chemical composition in various types of the alkaline granites. The chemical compositions of minerals from the granites are given in the tables in Supplementary material. The arfvedsonite contains 0.7–1.1 wt% TiO<sub>2</sub>, and the concentration of K<sub>2</sub>O varies from 1.4 to 1.7 wt%. There are also significant amounts of ZnO (0.2–2.9 wt%), MnO (0.6–1.5 wt%), SnO<sub>2</sub> (0.1–0.7 wt%), Li<sub>2</sub>O (0.5–0.9 wt%). We should also note a high content of F in amphiboles (up to F-arfvedsonite). Riebeckite is chemically different from arfvedsonite, mostly by significantly lower content of TiO<sub>2</sub> (< 0.3 wt%) and K<sub>2</sub>O (< 0.2 wt%), and very low concentrations of

#### Table 2

Major- and trace-element compositions of granites from the eastern segment of the Eastern Katugin massif.

Sample Granite	14216 Aeg	14217 Arf	14220 Aeg-Arf	14221 Aeg-Arf	14226 Arf	14231 Aeg-Arf	14232 Aeg-Arf	14233 Aeg-Arf	14234 Aeg-Arf
Location	Adit 1					Hole 107			
SiO- wt%	72 71	74 23	73 92	73.02	70.31	71.83	70 38	70.23	68 16
TiO <sub>2</sub>	0.87	0.19	0.05	0.07	0.07	0.25	0.22	0.28	0.23
Al <sub>2</sub> O <sub>3</sub>	8.33	10.69	10.05	11.23	11.16	9.70*	10.40*	10.50*	11.20*
Fe <sub>2</sub> O <sub>3</sub>	3.30	2.04	2.06	2.17	2.27	2.35	2.69	2.11	2.35
FeO	1.14	2.06	2.06	2.31	3.29	2.15	2.37	2.91	2.12
MnO	0.06	0.04	0.04	0.03	0.06	0.04	0.04	0.07	0.04
MgO	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02	0.05	0.10	0.09	0.07
CaO	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02	< 0.02
Na <sub>2</sub> O	3.43	4.01	5.41	4.53	5.56	5.89	6.20	5.38	6.22
K <sub>2</sub> O	3.87	4.11	3.8/	4.11	3.97	4.07	4.25	4.92	5.27
$P_2O_5$	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03
101	0.72	0.07	1.01	0.42	1 21	0.08	0.07	1.06	0.03
CO	< 0.06	< 0.06	< 0.06	< 0.06	< 0.06	< 0.06	0.75	< 0.06	< 0.06
F	0.70	0.78	1.73	0.34	1.40	2.50	2.48	1.87	1.60
-0	0.29	0.33	0.73	0.14	0.59				
D1	1076 15	1000.07	010.00	700 70	(00 54	1000 57	1170.00	1010.05	1154 41
RD, ppm	12/6.15	1090.87	810.86	/93./3	628.54	1292.57	11/8.00	1012.85	1154.41
SI V	24.12 625 70	04.85	9.12	1.72	20.25	14.40	55 71	12.72	45 30
Zr	27845 97	6570.14	2389 40	2562.07	8184 85	3073.01	3042.39	4346 71	4996 24
Nb	4190.89	2942.99	1255.81	1260.84	2287.48	4471.05	3108.15	3068.32	3202.29
Ва	119.66	223.03	19.45	32.33	62.88	30.23	46.68	35.47	31.98
La	159.38	136.88	63.69	50.26	82.03	208.05	145.47	141.64	169.63
Ce	479.99	457.10	202.05	165.61	282.05	667.10	482.76	452.00	527.70
Pr	65.65	56.39	23.93	19.15	35.56	83.42	58.49	60.03	65.93
Nd	257.28	189.98	78.24	55.61	121.74	280.43	196.55	223.60	228.16
Sm	104.30	43.10	17.47	12.74	26.62	67.61	44.67	69.95	54.45
Eu	4.36	1.50	0.63	0.39	0.85	2.40	1.53	2.78	1.81
Gd	101.64	32.45	13.65	8.62	16.36	52.23	31.34	61.15	34.68
1 D	24.41	5.44	2.42	1.59	2.44	9.26	5.13	9.78	5.05
Dy Ho	34 29	52.24 6 59	2.80	10.33	2 34	10.12	20.33	8 24	4.99
Fr	87 97	18.50	7.33	6.78	6.84	25.01	9.62	14 72	9.26
Tm	12.14	3.03	1.23	1.26	1.23	3.70	1.45	1.74	1.48
Yb	78.64	21.80	9.75	10.68	9.49	23.96	12.25	10.78	12.53
Lu	13.34	3.74	2.03	2.04	1.68	4.20	2.60	2.06	2.81
Hf	1031.88	239.93	90.06	91.95	336.04	117.40	103.03	152.90	180.32
Та	305.99	237.61	92.82	91.87	178.30	338.94	229.04	225.14	237.33
Th	277.11	63.24	21.58	31.96	40.76	82.36	43.40	36.34	43.88
U	121.94	108.66	42.17	39.38	100.02	156.70	103.99	109.21	110.75
A/NK	0.85	0.97	0.77	0.94	0.83	0.69	0.70	0.74	0.70
A51	0.85	0.97	0.77	0.94	0.83	0.69	0.70	0.74	0.70
MALL	7.20	1.00 8.12	0.28	1.00	0.52	0.99	10.45	10.98	11 40
(La/Yh)n	1.31	4.06	4.23	3.04	5.59	5.62	7 68	8 50	8 76
Eu/Eu <sup>*</sup>	0.13	0.12	0.13	0.11	0.13	0.12	0.13	0.13	0.13
Ce/Ce <sup>*</sup>	1.05	1.16	1.15	1.19	1.16	1.13	1.17	1.09	1.11
(Gd/Yb)n	1.09	1.26	1.19	0.68	1.46	1.84	2.16	4.80	2.34
Sample	14235	к 52/14	C-107-17	14237	14238	14240	14241	C-84-13	C-84-12
Granite	Aeg	Aeg	Aeg-Arf	Aeg-Arf	Arf	Aeg	Aeg-Arf	Aeg-Arf	Arf
	.0	-0	.0	-0		0	.0	.0	
Location	Hole 107			Hole 84					
SiO2,wt%	69.11	66.46	70.14	73.42	73.78	69.23	69.68	69.12	71.61
TiO <sub>2</sub>	0.30	0.25	0.41	0.11	0.10	0.29	0.37	0.30	0.29
$Al_2O_3$	9.90*	10.70	9.52	10.90	11.53	8.80*	9.80*	10.41	10.63
Fe <sub>2</sub> O <sub>3</sub>	3.20	3.70	3.87	2.14	1.55	3.78	2.01	5.65	1.81
FeO	0.79	2.30	1.29	2.47	2.99	0.99	3.21	0.90	3.77
MnO MaO	0.03	0.05	0.04	0.03	0.04	0.04	0.05	0.02	0.09
CaO	< 0.02	< 0.02	< 0.02	0.23	0.03	0.15	0.10	< 0.02	< 0.02
Na <sub>2</sub> O	4.61	6.55	4.96	5.09	4.75	6.10	5.42	6.80	<ul><li>5.47</li></ul>
K <sub>2</sub> O	4.77	3.63	3.56	4.30	4.47	3.60	4.28	3.58	3.63
$\tilde{P_2O_5}$	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03	< 0.03
$H_2O^-$	0.12	0.15	< 0.01	0.08	0.06	0.10	0.07	0.07	0.12
LOI	0.57	0.93	0.21	0.28	0.16	0.85	0.87	1.13	1.13
$CO_2$	< 0.06	< 0.06	0.07	< 0.06	< 0.06	< 0.06	< 0.06	0.10	0.19
F	0.68	2.90	1.35	1.10	0.60	2.70	2.00	2.48	2.00
-0	a	1.22	0.57	0.46	0.25			1.04	0.84
Rb, ppm	2661.54	1345.96	1351.91	522.38	398.70	1225.04	1331.07	1571.06	655.93
								(contin	ued on next page)

#### Table 2 (continued)

Sample Granite	14216 Aeg	14217 Arf	14220 Aeg-Arf	14221 Aeg-Arf	14226 Arf	14231 Aeg-Arf	14232 Aeg-Arf	14233 Aeg-Arf	14234 Aeg-Arf
Location	Adit 1					Hole 107			
Sr Y Zr	13.69 238.51 32431.20	15.69 195.00 22675.16	2.11 226.91 27901.51	30.72 239.24 302.15	14.42 512.56 1649.86	7.73 128.71 20610.03	12.43 131.67 12180.19	27.88 89.04 12506.29	35.60 67.60 10542.11
Nb Ba	3878.81 27.83	3036.38 8.46	3447.60 21.77	232.14 53.29	297.61 55.49	2737.92 30.60	3973.07 36.34	2899.85 40.40	3765.21 57.23
La Ce Pr	141.71 456.80 61.33	184.51 447.36 55.75	187.98 597.94 69.38	240.18 522.41 59.45	237.60 514.24 61.42	120.20 387.90 50.69	186.71 584.73 76.78	127.57 432.86 48.48	210.65 683.28 73.43
Nd Sm	224.69 70.37	204.68 49.25	252.37 79.76	222.47 56.46	227.83 61.00	179.24 49.56	281.41 79.87	167.49 43.69	241.89 55.94
Eu Gd Th	2.54 54.21 11.67	2.27 46.63 10.11	3.55 80.03 15.80	2.21 54.19 8.04	2.42 64.07 11.40	1.74 34.18 6.23	2.92 57.40 10.17	1.78 33.45 5.58	1.97 37.58 5.37
Dy Ho	76.89 17.02	71.10 14.35	96.25 18.71	47.57 9.63	73.77 16.83	42.28 9.61	54.91 9.71	36.07 7.13	32.16 5.72
Er Tm Vb	47.42 7.22 50.25	35.32 4.74 27.79	42.54 5.41 37.11	23.70 2.89 15.40	47.64 6.91 41.78	33.14 6.10 46.39	23.28 3.32 23.85	20.31 3.34 23.62	17.22 3.22 31.98
Lu Hf	9.47 1359.89	4.95 705.69	6.98 951.19	2.40 11.89	6.13 64.46	7.49 794.62	4.21 463.27	4.61 393.65	7.025 320.66
Ta Th	342.10 158.04	293.14 37.07	248.20 80.82	18.13 12.53	21.73 43.37	239.42 59.13	332.58 81.36	228.56 52.93	280.36 257.56
U A/NK ASI	0.78 0.78	0.73 0.73	152.82 0.79 0.79	7.08 0.84 0.82	0.91 0.89	0.63 0.63	0.72 0.72	0.69 0.69	0.82 0.82
f MALI	0.97 9.38	1.00 10.18	1.00 8.52	0.95 9.28	0.99 9.04	0.97 9.65	0.98 9.65	1.00 10.38	1.00 9.10
(La/Yb)n Eu/Eu <sup>*</sup> Ce/Ce <sup>*</sup>	1.82 0.13 1.09	4.30 0.15 0.98	3.28 0.14 1.17	0.12 0.97	3.68 0.12 0.95	1.68 0.13 1.11	5.06 0.13 1.09	3.49 0.14 1.23	4.26 0.13 1.22
(Gd/Yb)n Sample	0.91 14248	1.42 C-43-1	1.82 C-43-19	2.98 C-94-1	1.30 C-265-1	0.62 C-147-1	2.04 C-54-30	1.20 C-54-36	0.99 C-54-28
Granite	Aeg	Aeg-Arf	Aeg-Arf	Aeg-Arf	Aeg-Arf	Aeg-Arf	Arf	Arf	Arf
	** 1 40								
Location	Hole 43	70.00	71.40	Hole 94	Hole 265	Hole 147	Hole 54	<b>CO 10</b>	(0.71
Location SiO <sub>2</sub> ,wt% TiO <sub>2</sub>	Hole 43 71.59 0.35	72.68 0.29	71.49 0.35	Hole 94 68.92 0.22	Hole 265 66.94 0.23	Hole 147 67.82 0.28	Hole 54 67.38 0.22	69.48 0.22	69.71 0.18
Location SiO <sub>2</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub>	Hole 43 71.59 0.35 9.30 <sup>*</sup>	72.68 0.29 10.90	71.49 0.35 8.98	Hole 94 68.92 0.22 11.90*	Hole 265 66.94 0.23 11.70 <sup>*</sup>	Hole 147 67.82 0.28 11.20 <sup>*</sup>	Hole 54 67.38 0.22 11.10	69.48 0.22 11.00 <sup>*</sup>	69.71 0.18 10.67
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub>	Hole 43 71.59 0.35 9.30° 3.82	72.68 0.29 10.90 1.54	71.49 0.35 8.98 3.18	Hole 94 68.92 0.22 11.90 <sup>*</sup> 2.06	Hole 265 66.94 0.23 11.70° 3.52	Hole 147 67.82 0.28 11.20 <sup>*</sup> 3.63	Hole 54 67.38 0.22 11.10 2.34 2.34	69.48 0.22 11.00 <sup>°</sup> 1.98	69.71 0.18 10.67 1.74
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MCO	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.09	72.68 0.29 10.90 1.54 2.49	71.49 0.35 8.98 3.18 2.17	Hole 94 68.92 0.22 11.90 <sup>°</sup> 2.06 2.55 0.06	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.02	Hole 147 67.82 0.28 11.20 <sup>°</sup> 3.63 1.74 0.05	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06	69.48 0.22 11.00° 1.98 3.08	69.71 0.18 10.67 1.74 2.94
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO McO	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08	72.68 0.29 10.90 1.54 2.49 0.03	71.49 0.35 8.98 3.18 2.17 0.08	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02	Hole 147 67.82 0.28 11.20° 3.63 1.74 0.05	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02	69.48 0.22 11.00° 1.98 3.08 0.05	69.71 0.18 10.67 1.74 2.94 0.05
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> $A_{2}O_{3}$ Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06	72.68 0.29 10.90 1.54 2.49 0.03 < 0.02 0.37	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02	Hole 265 66.94 0.23 11.70 <sup>*</sup> 3.52 1.55 0.03 < 0.02 < 0.02	Hole 147 67.82 0.28 11.20° 3.63 1.74 0.05 < 0.02 < 0.02	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02	69.48 0.22 11.00° 1.98 3.08 0.05 < 0.02 < 0.02	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80	72.68 0.29 10.90 1.54 2.49 0.03 < 0.02 0.37 4.82	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02 6.66	Hole 265 66.94 0.23 11.70 <sup>*</sup> 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28	Hole 147 67.82 0.28 11.20° 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 6.73	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72	$69.48 \\ 0.22 \\ 11.00^{\circ} \\ 1.98 \\ 3.08 \\ 0.05 \\ < 0.02 \\ < 0.02 \\ < 0.02 \\ 6.84$	$69.71 \\ 0.18 \\ 10.67 \\ 1.74 \\ 2.94 \\ 0.05 \\ < 0.02 \\ < 0.02 \\ < 0.02 \\ 6.65 \\ \end{cases}$
Location $SiO_{2}$ ,wt% $TiO_{2}$ $Al_{2}O_{3}$ $Fe_{2}O_{3}$ FeO MnO MgO CaO $Na_{2}O$ $K_{2}O$	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71	Hole 94 68.92 0.22 $11.90^{\circ}$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52	Hole 147 67.82 0.28 11.20* 3.63 1.74 0.05 < 0.02 < 0.02 6.73 4.50	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26	$69.48 \\ 0.22 \\ 11.00^{\circ} \\ 1.98 \\ 3.08 \\ 0.05 \\ < 0.02 \\ < 0.02 \\ < 0.02 \\ 6.84 \\ 3.75$	$69.71 \\ 0.18 \\ 10.67 \\ 1.74 \\ 2.94 \\ 0.05 \\ < 0.02 \\ < 0.02 \\ 6.65 \\ 4.11$
Location SiO <sub>2</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O	Hole 43 71.59 0.35 9.30* 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ $	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85	Hole 94 68.92 0.22 11.90* 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05	Hole 147 67.82 0.28 11.20* 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16	$69.48 \\ 0.22 \\ 11.00^{*} \\ 1.98 \\ 3.08 \\ 0.05 \\ < 0.02 \\ < 0.02 \\ 6.84 \\ 3.75 \\ < 0.03 \\ 0.04$	$\begin{array}{c} 69.71\\ 0.18\\ 10.67\\ 1.74\\ 2.94\\ 0.05\\ < 0.02\\ < 0.02\\ 6.65\\ 4.11\\ < 0.03\\ 0.23\end{array}$
Location SiO <sub>2</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O Na <sub>2</sub> O Na <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92	Hole 147 67.82 0.28 11.20* 3.63 1.74 0.05 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07 1.00	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^*\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\end{array}$	$\begin{array}{c} 69.71\\ 0.18\\ 10.67\\ 1.74\\ 2.94\\ 0.05\\ < 0.02\\ < 0.02\\ 6.65\\ 4.11\\ < 0.03\\ 0.23\\ 1.61\end{array}$
Location SiO <sub>2</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O' LOI CO <sub>2</sub>	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 0.05 \\$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06	Hole 147 67.82 0.28 11.20* 3.63 1.74 0.05 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07 1.00 < 0.06	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^{*}\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07
Location $SiO_{2}$ ,wt% $TiO_2$ $Al_2O_3$ FeO MnO MgO CaO $Na_2O$ $K_2O$ $P_2O_5$ $H_2O^{-}$ LOI $CO_2$ F F	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ $	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$	Hole 94 68.92 0.22 $11.90^{*}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07 1.00 < 0.06 2.40	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^{*}\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25 \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05
Location $SiO_{2}$ ,wt% $TiO_2$ $Al_2O_3$ FeO MnO MgO CaO $Na_2O$ $K_2O$ $P_2O_5$ $H_2O^{-}$ LOI $CO_2$ F -O Rb, ppm	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ 1608.70$	71.49 0.35 8.98 3.18 2.17 0.08 $< 0.02< 0.02< 0.025.233.71< 0.031.850.67< 0.061.370.581676.45$	Hole 94 68.92 0.22 $11.90^{*}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^{*}\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25\\ \end{array}$	$\begin{array}{c} 69.71\\ 0.18\\ 10.67\\ 1.74\\ 2.94\\ 0.05\\ < 0.02\\ < 0.02\\ 6.65\\ 4.11\\ < 0.03\\ 0.23\\ 1.61\\ 0.07\\ 2.50\\ 1.05\\ 801.33\end{array}$
Location $SiO_{2}$ ,wt% $TiO_2$ $Al_2O_3$ FeO MnO MgO CaO $Na_2O$ $K_2O$ $P_2O_5$ $H_2O^{-}$ LOI $CO_2$ F -O Rb, ppm Sr	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ 1608.70 \\ 8.19 \\ \end{cases}$	71.49 0.35 8.98 3.18 2.17 0.08 $< 0.02< 0.025.233.71< 0.031.850.67< 0.061.370.581676.458.32$	Hole 94 68.92 0.22 $11.90^{\circ}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^{*}\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25\\ \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35
Location SiO <sub>22</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> Fe <sub>0</sub> MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48	72.68 0.29 10.90 1.54 2.49 0.03 $< 0.020.374.824.27< 0.03< 0.010.050.211.100.461608.708.19169.22$	71.49 0.35 8.98 3.18 2.17 0.08 $< 0.02< 0.025.233.71< 0.031.850.67< 0.061.370.581676.458.32325.70$	Hole 94 68.92 0.22 $11.90^{\circ}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 118.00	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^{*}\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25\\ \hline 706.76\\ 23.44\\ 165.85\\ \hline \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10
Location SiO <sub>22</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> Fe <sub>0</sub> MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454 90	72.68 $0.29$ $10.90$ $1.54$ $2.49$ $0.03$ $< 0.02$ $0.37$ $4.82$ $4.27$ $< 0.03$ $< 0.01$ $0.05$ $0.21$ $1.10$ $0.46$ $1608.70$ $8.19$ $169.22$ $13124.76$ $2607.22$	71.49 0.35 8.98 3.18 2.17 0.08 $< 0.02< 0.02< 0.025.233.71< 0.031.850.67< 0.061.370.581676.458.32325.7017105.762760.07$	Hole 94 68.92 0.22 $11.90^{*}$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 2026.02	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2691.75	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2129.60	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2450.10	69.48 0.22 $11.00^{\circ}$ 1.98 3.08 0.05 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2276.11	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2327.71
Location SiO <sub>22</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> Fe <sub>0</sub> MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ 1608.70 \\ 8.19 \\ 169.22 \\ 13124.76 \\ 3607.23 \\ 35.57 \\ \end{cases}$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21 59	Hole 94 68.92 0.22 $11.90^{*}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114 09	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17 $38$	69.48 0.22 $11.00^{*}$ 1.98 3.08 0.05 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01
Location $SiO_{22}wt\%$ $TiO_2$ $Al_2O_3$ $Fe_2O_3$ FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ \end{array}$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$ $1676.45$ $8.32$ $325.70$ $17105.76$ $3760.97$ $21.59$ $137.35$	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32	69.48 0.22 $11.00^{*}$ 1.98 3.08 0.05 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02
Location $SiO_{22}wt\%$ $TiO_2$ $Al_2O_3$ $Fe_2O_3$ FeO MnO MgO CaO $Na_2O$ $K_2O$ $P_2O_5$ $H_2O^{-}$ LOI $CO_2$ F -O Rb, ppm Sr Y Zr Nb Ba La Ce	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ \end{array}$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$ $1676.45$ $8.32$ $325.70$ $17105.76$ $3760.97$ $21.59$ $137.35$ $498.88$	Hole 94 68.92 0.22 $11.90^{*}$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 50.21	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^*\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ < 0.02\\ 6.84\\ 3.75\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25\\ \hline 706.76\\ 23.44\\ 165.85\\ 3090.62\\ 2376.11\\ 21.44\\ 178.61\\ 571.10\\ \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00
Location $SiO_{22}wt\%$ $TiO_2$ $Al_2O_3$ $Fe_2O_3$ FeO MnO MgO CaO $Na_2O$ $K_2O$ $P_2O_5$ $H_2O^{-}$ LOI $CO_2$ F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ \end{array}$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$ $1676.45$ $8.32$ $325.70$ $17105.76$ $3760.97$ $21.59$ $137.35$ $498.88$ $58.74$	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 50.21 58.84	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51	$\begin{array}{c} 69.48\\ 0.22\\ 11.00^*\\ 1.98\\ 3.08\\ 0.05\\ < 0.02\\ < 0.02\\ < 0.02\\ < 0.02\\ < 0.03\\ 0.04\\ 1.33\\ < 0.06\\ 3.25\\ \hline 706.76\\ 23.44\\ 165.85\\ 3090.62\\ 2376.11\\ 21.44\\ 178.61\\ 571.10\\ 65.32\\ \end{array}$	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87
Location SiO <sub>22</sub> ,wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ 253.13\\ \end{array}$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$ $1676.45$ $8.32$ $325.70$ $17105.76$ $3760.97$ $21.59$ $137.35$ $498.88$ $58.74$ $219.13$	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 503.21 58.84 211.50	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94	Hole 147 67.82 0.28 $11.20^{*}$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07	69.48 0.22 $11.00^{*}$ 1.98 3.08 0.05 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87 156.86
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78	72.68 0.29 10.90 1.54 2.49 0.03 < 0.02 0.37 4.82 4.27 < 0.03 < 0.01 0.05 0.21 1.10 0.46 1608.70 8.19 169.22 13124.76 3607.23 35.57 197.00 643.21 72.18 253.13 70.23	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 0.21	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 50.21 58.84 211.50 48.15 48.15	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21	Hole 147 67.82 0.28 $11.20^{*}$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72	69.48 0.22 $11.00^{*}$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87 156.86 35.58
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> F <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>•</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ 253.13\\ 70.23\\ 2.85\\ 50.62\\ \end{array}$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 69.00	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 50.21 58.84 211.50 48.15 1.60 2512	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 42.27	Hole 147 67.82 0.28 $11.20^{*}$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 22.52	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84	69.48 0.22 $11.00^{*}$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 42.45	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.02
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Ha Chan Ch	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14 76	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ 1608.70 \\ 8.19 \\ 169.22 \\ 13124.76 \\ 3607.23 \\ 35.57 \\ 197.00 \\ 643.21 \\ 72.18 \\ 253.13 \\ 70.23 \\ 2.85 \\ 59.63 \\ 11 \\ 45 \\ 150 \\ 1$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 68.00 13.80	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17	Hole 147 67.82 0.28 $11.20^{*}$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78	69.48 0.22 $11.00^*$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 742	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Tb Dy	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14.76 93.72	$72.68 \\ 0.29 \\ 10.90 \\ 1.54 \\ 2.49 \\ 0.03 \\ < 0.02 \\ 0.37 \\ 4.82 \\ 4.27 \\ < 0.03 \\ < 0.01 \\ 0.05 \\ 0.21 \\ 1.10 \\ 0.46 \\ 1608.70 \\ 8.19 \\ 169.22 \\ 13124.76 \\ 3607.23 \\ 35.57 \\ 197.00 \\ 643.21 \\ 72.18 \\ 253.13 \\ 70.23 \\ 2.85 \\ 59.63 \\ 11.45 \\ 72.86 \\ \end{cases}$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 68.00 13.80 94.85	Hole 94 68.92 0.22 $11.90^*$ 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99 27.91	Hole 265 66.94 0.23 11.70* 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17 61.56	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43 44.34	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 < 0.02 6.72 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78 41.13	69.48 0.22 $11.00^*$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 7.42 45.71	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 < 0.02 < 0.02 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 2203.17 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.92 4.11 24.60
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O <sup>-</sup> LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Tb Dy Ho	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14.76 93.72 18.49	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ 253.13\\ 70.23\\ 2.85\\ 59.63\\ 11.45\\ 72.86\\ 12.46\\ \end{array}$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 68.00 13.80 94.85 18.83	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99 27.91 4.27	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17 61.56 13.46	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 6.73 4.50 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43 44.34 9.68	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78 41.13 7.86	69.48 0.22 $11.00^{\circ}$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 7.42 45.71 9.05	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11 24.60 4.20
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Tb Dy Ho Er	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14.76 93.72 18.49 40.65	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ 253.13\\ 70.23\\ 2.85\\ 59.63\\ 11.45\\ 72.86\\ 12.46\\ 29.69\\ \end{array}$	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 68.00 13.80 94.85 18.83 53.68	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 48.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99 27.91 4.27 9.21	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17 61.56 13.46 33.70	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43 44.34 9.68 28.32	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78 41.13 7.86 20.47	69.48 0.22 $11.00^{\circ}$ 1.98 3.08 0.05 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 7.42 45.71 9.05 21.61	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11 24.92 4.11 24.60 4.20 9.81
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Tb Dy Ho Er Tm	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14.76 93.72 18.49 40.65 4.97	$\begin{array}{c} 72.68\\ 0.29\\ 10.90\\ 1.54\\ 2.49\\ 0.03\\ < 0.02\\ 0.37\\ 4.82\\ 4.27\\ < 0.03\\ < 0.01\\ 0.05\\ 0.21\\ 1.10\\ 0.46\\ 1608.70\\ 8.19\\ 169.22\\ 13124.76\\ 3607.23\\ 35.57\\ 197.00\\ 643.21\\ 72.18\\ 253.13\\ 70.23\\ 2.85\\ 59.63\\ 11.45\\ 72.86\\ 12.46\\ 29.69\\ 3.75\\ \end{array}$	71.49 $0.35$ $8.98$ $3.18$ $2.17$ $0.08$ $< 0.02$ $< 0.02$ $5.23$ $3.71$ $< 0.03$ $1.85$ $0.67$ $< 0.06$ $1.37$ $0.58$ $1676.45$ $8.32$ $325.70$ $17105.76$ $3760.97$ $21.59$ $137.35$ $498.88$ $58.74$ $219.13$ $69.22$ $2.96$ $68.00$ $13.80$ $94.85$ $18.83$ $53.68$ $6.92$	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 488.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99 27.91 4.27 9.21 1.15	Hole 265 66.94 0.23 $11.70^{\circ}$ 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17 61.56 13.46 33.70 4.73	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43 44.34 9.68 28.32 5.30	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78 41.13 7.86 20.47 2.87	69.48 0.22 $11.00^{\circ}$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 7.42 45.71 9.05 21.61 2.86	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11 24.92 4.11 24.60 4.20 9.81 1.36
Location SiO <sub>22</sub> wt% TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO Na <sub>2</sub> O K <sub>2</sub> O P <sub>2</sub> O <sub>5</sub> H <sub>2</sub> O LOI CO <sub>2</sub> F -O Rb, ppm Sr Y Zr Nb Ba La Ce Pr Nd Sm Eu Gd Tb Dy Ho Er Tm Yb	Hole 43 71.59 0.35 9.30° 3.82 1.58 0.08 < 0.02 0.06 5.80 4.06 < 0.03 0.04 0.98 < 0.06 1.87 1395.00 14.58 253.48 1597.18 4454.89 35.72 171.89 539.42 72.17 266.20 85.78 3.29 68.32 14.76 93.72 18.49 40.65 4.97 27.95 50 50 50 50 50 50 50 50 50 5	72.68 0.29 10.90 1.54 2.49 0.03 < 0.02 0.37 4.82 4.27 < 0.03 < 0.01 0.05 0.21 1.10 0.46 1608.70 8.19 169.22 13124.76 3607.23 35.57 197.00 643.21 72.18 253.13 70.23 2.85 59.63 11.45 72.86 12.46 29.69 3.75 19.62 19.62	71.49 0.35 8.98 3.18 2.17 0.08 < 0.02 < 0.02 5.23 3.71 < 0.03 1.85 0.67 < 0.06 1.37 0.58 1676.45 8.32 325.70 17105.76 3760.97 21.59 137.35 498.88 58.74 219.13 69.22 2.96 68.00 13.80 94.85 18.83 53.68 6.92 42.87	Hole 94 68.92 0.22 11.90° 2.06 2.55 0.06 < 0.02 < 0.02 6.66 4.53 < 0.03 0.07 0.96 < 0.06 2.50 1157.19 24.50 57.08 48.05 3026.93 23.42 142.98 503.21 58.84 211.50 48.15 1.60 35.12 4.99 27.91 4.27 9.21 1.15 7.27 9.51	Hole 265 66.94 0.23 11.70° 3.52 1.55 0.03 < 0.02 < 0.02 < 0.02 7.28 4.52 < 0.03 0.05 0.92 < 0.06 3.30 1517.72 43.47 245.64 4371.85 2681.75 114.09 134.27 457.87 56.02 202.94 56.21 2.05 43.27 9.17 61.56 13.46 33.70 4.73 29.21 20.5	Hole 147 67.82 0.28 $11.20^*$ 3.63 1.74 0.05 < 0.02 < 0.02 < 0.02 < 0.03 0.07 1.00 < 0.06 2.40 1783.84 37.24 126.55 7304.63 2183.60 14.67 98.04 339.46 42.23 155.79 45.05 1.57 33.53 6.43 44.34 9.68 28.32 5.30 42.08 14.67	Hole 54 67.38 0.22 11.10 2.34 3.17 0.06 < 0.02 < 0.02 < 0.02 4.26 < 0.03 0.16 1.57 0.19 2.72 1.15 819.48 16.00 118.00 3934.43 2459.10 17.38 196.32 616.88 69.51 238.07 52.72 1.88 40.84 6.78 41.13 7.86 20.47 2.87 17.97	69.48 0.22 $11.00^{\circ}$ 1.98 3.08 0.05 < 0.02 < 0.02 < 0.02 6.84 3.75 < 0.03 0.04 1.33 < 0.06 3.25 706.76 23.44 165.85 3090.62 2376.11 21.44 178.61 571.10 65.32 235.53 54.24 2.06 43.45 7.42 45.71 9.05 21.61 2.86 17.35 2.52	69.71 0.18 10.67 1.74 2.94 0.05 < 0.02 < 0.02 6.65 4.11 < 0.03 0.23 1.61 0.07 2.50 1.05 801.33 13.35 52.10 2377.71 22.01 128.02 423.00 47.87 156.86 35.58 1.11 24.92 4.11 24.92 4.11 24.60 4.20 9.81 1.36 9.51 1.51

#### Table 2 (continued)

Sample Granite	14216 Aeg	14217 Arf	14220 Aeg-Arf	14221 Aeg-Arf	14226 Arf	14231 Aeg-Arf	14232 Aeg-Arf	14233 Aeg-Arf	14234 Aeg-Arf
Location	Adit 1					Hole 107			
Hf	62.88	427.07	498.52	18.44	143.48	224.01	133.47	106.71	77.84
Та	308.91	219.46	276.09	249.76	214.18	175.55	199.17	196.65	180.01
Th	255.27	131.47	215.34	43.87	83.02	64.38	50.90	34.97	24.18
U	145.84	127.08	119.83	110.21	99.89	83.87	114.69	95.52	80.62
A/NK	0.67	0.87	0.71	0.75	0.69	0.70	0.71	0.72	0.69
ASI	0.66	0.82	0.71	0.75	0.69	0.70	0.71	0.72	0.69
Fe index	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00	1.00
MALI	9.80	8.72	8.94	11.19	11.80	11.23	10.98	10.58	10.76
(La/Yb) <sub>N</sub>	3.98	6.50	2.07	12.73	2.97	1.51	7.07	6.66	8.71
Eu/Eu*	0.13	0.14	0.13	0.12	0.13	0.12	0.12	0.13	0.11
Ce/Ce*	1.08	1.20	1.24	1.22	1.18	1.18	1.18	1.18	1.20
(Gd/Yb) <sub>N</sub>	2.07	2.57	1.34	4.09	1.25	0.67	1.92	2.12	2.22

Note: Aeg - aegirine granite; Aeg-Arf - aegirine-arfvedsonite granite; Arf - arfvedsonite granite.

 $A/NK = molar Al_2O_3/(Na_2O + K_2O); ASI = molar Al_2O_3/(CaO + Na_2O + K_2O); Fe index = FeO^*/(FeO^* + MgO).$ 

 $\text{FeO}^* = \text{FeO} + 0.8998 \text{ } x \text{ } \text{Fe}_2\text{O}_3$ ; MALI = Na<sub>2</sub>O + K<sub>2</sub>O - CaO; Eu/Eu<sup>\*</sup> = Eu<sub>N</sub>/ $\vee$ (Sm<sub>N</sub> x Gd<sub>N</sub>); Ce/Ce<sup>\*</sup> = Ce<sub>N</sub>/ $\vee$ (La<sub>N</sub> x Pr<sub>N</sub>); N, chondrite-normalized (Wakita et al., 1970). \*, Al<sub>2</sub>O<sub>3</sub> analysed by atomic absorption spectroscopy.

#### MnO and F.

Apart from an unusual emerald-green color, aegirine has a practically ideal composition – the content of  $Al_2O_3$  does not exceed 1.3 wt%, and the content of CaO is less than 0.1 wt%. Only  $SnO_2$  (up to 1.6 wt%) is present as a significant impurity.

Unlike the above mentioned minerals, astrophyllite may contain significant amounts of ore components, e.g. the content of  $ZrO_2$  reaches 1.0 wt%,  $Nb_2O_5$  is up to 5.0 wt%,  $Ta_2O_5 -$  up to 0.5 wt%. In addition, there are high concentrations of MnO (up to 2.7 wt%),  $Rb_2O$  (up to 2.0 wt%) and  $SnO_2$  (up to 2.1 wt%).

### 5. Geochemistry of the Katugin granites from the eastern segment of the Eastern Katugin massif

The chemical composition of representative samples of Arf, Aeg-Arf and Aeg granites of the eastern segment of the Eastern Katugin massif are summarized in Table 2. All granites demonstrate elevated and high alkalinity (Na<sub>2</sub>O +  $K_2O = 7.3-11.8$  wt%). The silica content in Arf, Aeg-Arf and Aeg granites varies from 66 to 74 wt%, which classifies these rocks in the TAS diagram (Sharpenok et al., 2013) as alkaline quartz syenites, alkaline granites and granites (Fig. 3a). The granites have A/NK [molar  $Al_2O_3/(Na_2O + K_2O)$ ] = 0.63–0.97 and ASI [molar  $Al_2O_3/(CaO + Na_2O + K_2O)] = 0.63-0.97$ , which classifies them as peralkaline granites (Fig. 3b). According to Frost et al. (2001) all granites are ferroan (Fe index = 0.95-1.00), alkali and alkali-calcic (MALI  $[Na_2O + K_2O-CaO] = 7.3-11.8$ ). Fig. 3a-d show the decreasing of the contents of  $Na_2O + K_2O$  and of the MALI index, and some increase of the A/NK index with increasing of the SiO<sub>2</sub> content. The granites also have extremely low CaO contents (up to 0.37 ppm), MgO (up to 0.23 ppm), and  $P_2O_5$  ( < 0.03 ppm). On the other hand, the Arf, Aeg-Arf and Aeg granites are strongly enriched in F (0.34–3.30 wt%). Besides, there is a tendency for the Arf, Aeg-Arf and Aeg granites of the increasing of the Na<sub>2</sub>O content with increased concentration of F, which is probably related to the amount of cryolite in the rock (Fig. 3e). There is also a positive correlation between Na2O and Fe2O3tot, which is typical for the fractionated peralkaline silicic rocks (Scaillet and Macdonald, 2003; Costi et al., 2009) (Fig. 3f).

All granites have high concentrations of most incompatible elements (Rb, Y, Zr, Hf, Nb, Ta, Th, U, REEs), except for Eu. They are also characterized by low Sr and Ba concentrations. The spider diagram (Fig. 4a) shows the well-developed negative Ba, Sr, Eu, and Ti as well as positive Rb, Th, U, Nb, Ta, Zr, and Hf anomalies. Notably, the Arf, Aeg-Arf and Aeg granites have high but variable abundances of HFSE and REE, which is probably related to a random distribution of Zr-Hf, Nb-Ta and REE-bearing minerals in the granites. Extremely high contents of Zr

(up to 32431 ppm), Hf (up to 1360 ppm), Nb (4471 ppm), Ta (339 ppm), total REE (up to 1584 ppm) occur in the Ae and Ae-Arf granites. The REE distribution in the granites is characterized by weakly and moderate enrichment in LREE ( $[La/Yb]_N = 1.3-12.7$ ) often with a slightly positive Ce anomaly (Ce/Ce<sup>\*</sup> = 0.95–1.24) and always with a negative Eu anomalies (Eu/Eu<sup>\*</sup> = 0.11–0.15) (Fig. 4b). The HREE segment demonstrate as positive as and negative slopes with [Gd/Yb]\_N = 0.62–4.80 (Fig. 4b).

The high concentrations of Rb with respect to those of Sr and Ba (Rb/Sr = 17–640; Rb/Ba = 5–159) are indicative the granite formation resulted from a deep differentiation of the parental melts. According to Frost and Frost (2008, 2011), Aeg, Aeg-Arf and Arf granites are peralkaline ferroan alkalic and alkali-calcic rocks, so they can be considered as differentiates of transitional in composition or alkali basaltic magma affected by some crustal contamination. Consequently a basaltic magma of transitional composition or OIB type can be considered as a source for Aeg, Aeg-Arf and Arf granites. Increasing of the A/NK index and decreasing of the MALI index with increased SiO<sub>2</sub>, (Fig. 3c and d) – all these features suggest some contamination of the peralkaline differentiates by crustal melts during the formation of enriched SiO<sub>2</sub> granites. Overlapping of the compositions of Aeg, Aeg-Arf and Arf granites in all diagrams and similar trends are indicative for the same source for the granites.

Table 3 presents the results of Sm–Nd isotopic–geochemical investigations of the Arf, Aeg-Arf and Aeg granites as well as the ore segregations. They are characterized by negative and CHUR-close  $\epsilon$ Nd (*t*) values of 0.0...–1.9 (Fig. 5) and usually by relatively high <sup>1475</sup>m/<sup>144</sup>Nd values. Therefore, the Nd model ages can be calculated only for two samples of granites (T<sub>Nd</sub>(*DM*) = 2.6–2.7 Ga). The two-stage Nd model age values (T<sub>Nd</sub>(*DM*-2st)) of all samples are estimated between 2.5 and 2.7 Ga. The Aeg-Arf granite with SiO<sub>2</sub> = 74.2 wt% (maximum contents – see Table 3) has the lowest  $\epsilon$ Nd(*t*) value of – 1.9, which confirms our conclusion about the role of crustal contamination of the peralkaline differentiates during the formation of silica-enriched granites. Similar  $\epsilon$ Nd(*t*) values and similar Nd model ages of the Arf, Aeg-Arf and Aeg granites and ore segregations suggest their common source.

#### 6. The mineralization of the Katugin deposit

Most of the ore bodies located in the eastern segment of the Eastern Katugin intrusive massif. They are represented by the Arf, Aeg-Arf and Aeg granites, enriched in ore minerals, or by veins or lens-like bodies of the cryolite-rich rocks. The results of detailed mineralogical study suggest three major types of mineralization in the Katugin deposit. The



Fig. 3. (Na<sub>2</sub>O + K<sub>2</sub>O)–SiO<sub>2</sub> diagram of Sharpenok et al. (2013) (a), A/NK [molar Al<sub>2</sub>O<sub>3</sub>/(Na<sub>2</sub>O + K<sub>2</sub>O)] – ASI [molar Al<sub>2</sub>O<sub>3</sub>/(CaO + Na<sub>2</sub>O + K<sub>2</sub>O)] diagram (b), MALI [Na<sub>2</sub>O + K<sub>2</sub>O – CaO] – SiO<sub>2</sub> diagram of Frost et al. (2001) (c), A/NK – SiO<sub>2</sub> diagram (d), F – Na<sub>2</sub>O diagram (e), Fe<sub>2</sub>O<sub>3</sub>Tot – Na<sub>2</sub>O diagram (f) for Arf, Aeg-Arf and Aeg granites of the eastern segment of the Eastern Katugin massif.

chemical compositions of ore minerals from the granites are provided in the Supplementary material.

#### 6.1. Zr mineralization

Zircon is present in all rocks of the Eastern Katugin massif, but its amount varies from single crystals to the nest aggregates, containing up to 15–20% of the rock volume. Size of the crystals varies strongly from D.P. Gladkochub et al.



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Fig. 4. Multi-element diagrams normalized to primitive mantle composition (Sun and McDonough, 1989) (a) and REE diagrams normalized to chondrite (Wakita et al., 1970) (b) for Arf, Aeg-Arf and Aeg granites of the eastern segment of the Eastern Katugin massif.

few micrometers to 1.5 cm (Arkhangel'skaya et al., 2012). Zircons in some cases are free of mineral inclusions, but in many cases contain tiny (2–15  $\mu$ m) inclusions – predominantly quartz, bastnaesite, fluocerite, gagarinite, tveitite, ittrofluorite, thorite and cryolite (Fig. 6a and b). We suggest their syn-crystallization nature.

#### 6.2. Ta-Nb-REE mineralization

Ta-Nb-REE minerals are present as: a) individual grains and aggregates in granites; b) inclusions in zircons (see above); c) interstitial aggregates among silicate minerals. Pyrochlore usually occurs as individual crystals or their aggregates and in most cases contains mineral inclusions and opaque rim (Fig. 6c). In most cases the crystals of pyrochlore have complex structure caused by variations in chemical composition or by inclusions of other minerals. Bastnaesite, columbite, fluocerite as well as hydrated pyrochlore are present in marginal parts of the crystals. Ore-mineral aggregates are usually composed of pyrochlore, columbite, fluocerite, ilmenite and sometimes sulphides (sphalerite, galena, pyrrotyte or pyrite) (Fig. 6d). Interstitial segregations consist of fluorides and fluocarbonates, sometimes with small amount of chlotite. Tveitite with numerous oriented inclusions of fluocerite or early bastnaesite, predominates (Fig. 6e and f). Late bastnaesite usually

Table 3 Sm–Nd isotopic data for granites and ore segregations the Eastern Katugin massif.

13023Arf (72.9)206630.1143.90.12630.511656 $\pm 8$ $-0.5$ 2583254814382Aeg-Arf (67.6)206663.1234.00.16300.512156 $\pm 5$ $-0.4$ 2547C-6-1Arf (72.7)206632.9128.10.15510.512044 $\pm 2$ $-0.5$ 2554C-15-1Aeg-Arf (74.2)206678.5167.40.28370.513723 $\pm 2$ $-1.9$ 2667C-38A-1Aeg-Arf (72.7)206673.2195.40.22660.513014 $\pm 2$ $-0.6$ 2558C-54-36Arf (69.5)206655.5216.00.15570.511994 $\pm 2$ $-1.7$ 2648C-94-1Aeg (68.9)206652.6219.00.14520.511907 $\pm 2$ $-0.6$ 2559C-54-27Ore (64.6)2066153.7671.00.13850.511780 $\pm 3$ $-1.3$ 27512616	Sample	Rock type (SiO <sub>2</sub> , wt%)	T, Ma	Sm (ppm)	Nd (ppm)	<sup>147</sup> Sm/ <sup>144</sup> Nd	$^{143}\text{Nd}/^{144}\text{Nd}~\pm~2\sigma$	$\epsilon_{\rm Nd}({\rm T})$	T <sub>Nd(DM)</sub>	T <sub>Nd(DM-2st)</sub>
(-121-1) Ore $(64.7)$ 2066 184.9 642.0 0.1740 0.512331 + 2 0.0 2507	13023 14382 C-6-1 C-15-1 C-38A-1 C-54-36 C-94-1 C-54-27 C-121-1	Arf (72.9) Aeg-Arf (67.6) Arf (72.7) Aeg-Arf (74.2) Aeg-Arf (72.7) Arf (69.5) Aeg (68.9) Ore (64.6) Ore (64.7)	2066 2066 2066 2066 2066 2066 2066 2066	30.1 63.1 32.9 78.5 73.2 55.5 52.6 153.7 184.9	143.9 234.0 128.1 167.4 195.4 216.0 219.0 671.0 642.0	0.1263 0.1630 0.1551 0.2837 0.2266 0.1557 0.1452 0.1385 0.1740	$\begin{array}{rrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrrr$	-0.5 -0.4 -0.5 -1.9 -0.6 -1.7 -0.6 -1.3 0.0	2583 2751	2548 2547 2554 2667 2558 2648 2559 2616 2507

Note: Rock types are Arf (arfvedsonite granite), Aeg-Arf (aegirine-arfvedsonite granite), Aeg (aegirine granite), Ore (ore segregation).



Fig. 5. The  $\epsilon Nd(t)$  – age diagram for Arf, Aeg-Arf and Aeg granites and the ore segregations of the eastern segment of the Eastern Katugin massif.

develops after fluorides, composing small veins and individual aggregates, sometimes – symplectites with chlorite.

The composition of pyrochlores varies significantly even within a single grain in all rock types. The content of microlite (#Ta) in pyrochlores is relatively small – about 1–4%. Maximum variations are found for uranium (1–12 wt% UO<sub>2</sub>) and lead (0–24 wt% PbO). Small amounts of Ta (0.5 wt% Ta<sub>2</sub>O<sub>5</sub>) are found in columbite. Only light REE are present in significant amounts in bastnaesites, but small amounts of thorium are also found (up to 3 wt% ThO<sub>2</sub>).

Fluorides (fluocerite, tveitite, gagarinite, Y-fluorite) occur as clusters with or without cryolites in granites (see below), but sometimes as inclusions in zircons. Tveitite show largest variations in chemical composition. Apart from a normal tveitite some Ba-tveitite is also found. Significant variations of the light/heavy (+Y) REE ratio are observed in gagarinite. The sum of light REE in gagarinites varies from 0 to 20 wt%. Only light REE and Th in detectable amounts are found in fluocerite.

#### 6.3. Aluminum fluoride mineralization

Cryolite is a dominating mineral among fluoroaluminates of the Katugin deposit. Cryolite commonly forms small isolations in granites and larger veins or lens-like bodies of cryolite-rich (> 30–50 vol%) rocks (Arkhangel'skaya et al., 1993, 2012). This mineral associates with other fluoroaluminates and fluorides (weberite Na<sub>2</sub>MgAlF<sub>7</sub>, chiolite Na<sub>5</sub>Al<sub>3</sub>F<sub>14</sub>, neighborite NaMgF<sub>3</sub>, fluorite, tveitite-(Y), fluocerite-(Ce),

gagarinite-(Y), ittrofluorite, elpasolite K<sub>2</sub>NaAlF<sub>6</sub>, simmonsite Na<sub>2</sub>LiAlF<sub>6</sub>) and products of their alteration (gearksutite CaAl-F<sub>4</sub>(OH)·H<sub>2</sub>O, prosopite CaAl<sub>2</sub>(F,OH)<sub>8</sub>, thomsenolite NaCaAlF<sub>6</sub>·H<sub>2</sub>O, pachnolite NaCaAlF<sub>6</sub>·H<sub>2</sub>O, ralstonite Na<sub>x</sub>Mg<sub>x</sub>Al<sub>2-x</sub>(F,OH)<sub>6</sub>·H<sub>2</sub>O) (Bykov and Arkhangel'skaya, 1995; Arkhangel'skaya et al., 2012; Sharygin and Vladykin, 2014).

#### 7. The age of the Katugin deposit origin

Ages of the host rocks and of the mineralization are of a crucial importance for the modelling of any mineral deposit. The crystallization age of the Aeg-Arf granite (Eastern Katugin massif) has been reported at 2066  $\pm$  6 Ma (U–Pb TIMS zircon dating; Larin et al., 2002). Our new U–Pb ages of zircons extracted from the ore segregation in the Aeg-Arf granite of this massif (sample KTG-241) provide the time of the ore formation. The ore segregations in the studied granites are composed mostly of zircon and pyrochlore (Arkhangel'skaya et al., 1993).

The zircon grains extracted from the sample are mainly transparent to semitransparent, euhedral, occasionally subhedral (Fig. 7I–IV), from cherry to dark cherry and light pink colored. Rare colorless euhedral bipyramidal crystals are also present. In the transmitted light and in the CL images, the zircons demonstrate zoned cores and homogeneous rims (Fig. 7V–X) with a high intensity of luminescence (Fig. 7V). The cores are enriched in mineral, primary melt, gaseous, and fluid inclusions, while the rims are devoid of melt inclusions (Fig. 7VI–VIII). Fluid and gaseous inclusions are located predominantly at the core–rim boundaries (Fig. 7IV–IX) as well as in fractures penetrating both cores and rims (Fig. 7X). The results of thermobarogeochemical studies demonstrate that the homogenization temperature of melt inclusions in zircon is 760–780 °C.

For geochronological study, the most transparent and homogeneous single zircon grains of cherry and dark cherry colors were selected from size fractions 100-150 and 200 µm (Nos. 1-4, Table 4), and one colorless grain from size fraction 200 µm (No. 4, Table 4). Two of these zircon grains were air-abraded (Nos. 2 and 4, Table 4) with up to 20% of the grain being removed. In the concordia plot (Fig. 8), the data points form a discordia line with the upper intercept age of 2055  $\pm$  7 Ma and the lower one corresponding to 973  $\pm$  110 Ma (MSWD = 1.3). The morphological features of the zircons from the ore segregation in the Aeg-Arf granite of the Eastern Katugin massif containing primary melt inclusions suggest their crystallization from a melt. In considering the good fit of the <sup>207</sup>Pb/<sup>206</sup>Pb ages obtained from both air-abraded and non-treated zircons, it is possible to conclude that the growth of the rims in the studied zircons is not related to any overlapping process, and reflect the time of crystallization of the melt, parental to the ore segregations.

The obtained age of 2055  $\pm$  7 Ma for the zircons from the ore segregation coincides within the error limits with the estimated age of the crystallization of the Katugin Aeg-Arf granites (2066  $\pm$  6 Ma, U–Pb zircon age, Larin et al., 2002). Notably the accessory zircons from the granites contain numerous inclusions of minerals of rare and REE



Fig. 6. Photomicrograph in plane polarized light (a and b) and backscattered electron images (c-f) of some ore minerals of the Katugin ore deposit. (a and b) – inclusion-rich and inclusion-free zircons in Aeg-Arf granite; (c) – pyrochlore grains rimmed by bastnaesite; (d) – ore-mineral aggregate, composed of pyrochlore, columbite, ilmenite and sphalerite; (e) – interstitial segregation, consisting of fluorides and fluocarbonates with small amount of chlorite; (f) – fragment of (e). Abbreviations: Bs – bastnaesite, Col – columbite, Ilm – ilmenite, Pcl – pyrochlore, Sp – sphalerite, Zrn – zircon, Qz – quartz, Kfs – K-feldspar, Ab – albite, 1 – tveitite with numerous fine oriented inclusions of early bastnaesite, 2 – tveitite with numerous fine oriented inclusions of fluocerite, 3 – late bastnaesite, 4 – chlorite with inclusions of late bastnaesite.

elements (bastnaesite, fluorite, Y–fluorite, fluocerite, monazite, thorite, synchisite, gagarinite, cryolite and others) (Levashova et al., 2014) and the same minerals often occur in ore segregations. We assume that formation of the Katugin deposit is genetically related to processes of formation of the Katugin alkaline granites.

#### 8. Discussion

#### 8.1. Magmatic nature of the mineralization of the Katugin deposit

Geological observations, geochemical characteristics of the granites and results of mineralogical, thermobarogeochemical and geochronological studies provide important evidence in favour of the magmatic models for the generation of rare-metal (and REE) enriched granite and, D.P. Gladkochub et al.

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Fig. 7. Scanning electron microscope images of the zircon crystals from ore segregation (sample KTG-241) within Aeg-Arf granite: secondary electron – I–IV; cathodoluminescence – V; transmitted light optic microscope – VI–X. C – core; R – rim; FI – fluid inclusions; MI – melt inclusions.

consequently, for the genesis of the Katugin deposit, and allow to dismiss the hypothesis of the metasomatic origin of this deposit. We conclude that most of mineralization of the Katugin ore deposit occurred during the magmatic stage of evolution of the alkaline granitic melt, which had been a source for the Arf, Aeg-Arf and Aeg granites. The following lines of evidence support our conclusion.

1. Petrographic data clearly indicate that the ore minerals crystallized from the silicate melt during the magmatic stage. This is obvious for zircon, pyrochlore, gagarinite, ittrofluorite (both individual crystals and their monomineral aggregates), as well as for the polymineral aggregates of fergusonite, pyrochlore, fluocerite, ilmenite and sulphides. For example, zircons are found as inclusions in major minerals (e.g. albite, potassium feldspar and quartz) and pyrochlores sometimes occur as inclusions in albite. This is characteristic for the early magmatic stage of ore mineralization. Besides most of pyrochlore and bastnaesite crystals in granites contain high amount of F (3.32–5.50 and 8.13–8.64 wt%, correspondingly), which is also indicative for their crystallization in the magmatic stage (Yin et al., 2013; Huang et al., 2014).

- 2. Petrographic data show that the accessory cryolites in granites have been crystallized from the magmatic silicate melt enriched with fluorine. However, cryolites in larger veins and lens-like bodies have been crystallized at the latest stage from the fluorine salt melt (see below).
- 3. Thermobarogeochemical studies demonstrated that both rock-

#### Table 4

U-Pb isotopic data for zircons from the ore segregation in the Katugin aegirine-arfvedsonite granite (sample KTG-241).

No.	Grain size (µm) and their	U/Pb*	Isotopic ratios						Age, Ma		
	characteristics		<sup>206</sup> Pb/ <sup>204</sup> Pb	<sup>207</sup> Pb/ <sup>206</sup> Pb <sup>a</sup>	<sup>208</sup> Pb/ <sup>206</sup> Pb <sup>a</sup>	<sup>207</sup> Pb/ <sup>235</sup> U	<sup>206</sup> Pb/ <sup>238</sup> U		<sup>207</sup> Pb/ <sup>235</sup> U	<sup>206</sup> Pb/ <sup>238</sup> U	<sup>207</sup> Pb/ <sup>206</sup> Pb
1 2	100–150, 1 grn, dark cherry-colored > 200, <i>A</i> = 20%, 2 grn, cherry-	1.84 1.74	257 247	$0.1227 \pm 3$ $0.1254 \pm 2$	$0.5928 \pm 1$ $0.4684 \pm 1$	$5.7273 \pm 211$ $6.1623 \pm 298$	$0.3384 \pm 7$ $0.3565 \pm 16$	0.84 0.70	$1936 \pm 7$ $1999 \pm 9$	$1879 \pm 6$ $1965 \pm 8$	$1997 \pm 4$ 2034 ± 3
3 4	colored $> 200, 1$ grn, cherry-colored $> 200, A = 20\%, 2$ grn, cherry-	1.96 1.94	301 487	$0.1252 \pm 4$ $0.1257 \pm 2$	$0.3296 \pm 1$ $0.3764 \pm 1$	$6.2436 \pm 250$ $6.3099 \pm 189$	$0.3618 \pm 7$ $0.3640 \pm 11$	0.51 0.98	$2011 \pm 8$ $2020 \pm 6$	$1991 \pm 4$ 2001 ± 6	$2031 \pm 6$ $2039 \pm 2$
5	> 200, 1 grn, colorless	2.13	792	$0.1256~\pm~3$	$0.3553~\pm~1$	6.3254 ± 259	0.3654 ± 11	0.78	$2022~\pm~8$	$2008~\pm~6$	$2037~\pm~5$

*Note*: <sup>a</sup>, isotope ratios are corrected for blank and common lead; Rho, coefficient of correlation of <sup>207</sup>Pb/<sup>235</sup>U-<sup>206</sup>Pb/<sup>235</sup>U; A%, amount of matter removed by air-abrasion treatment; 2 grn, number of zircon grains used for U–Pb analysis; <sup>\*</sup>, sample not weighed. Errors are 2-sigma and correspond to the last significant digits after the decimal point.



**Fig. 8.** Concordia diagram showing analyses of zircon from sample KTG-241. Spot numbers correspond to those in Table 4.

forming minerals of the granites (e.g. albite, quartz, and arfvedsonite) and accessory minerals (e.g. zircon and pyrochlore) contain syngenetic (primary) melt inclusions. These results demonstrate that the homogenization temperature of melt inclusions in zircon is 760–780 °C. Besides large salt-fluid inclusions (30–50  $\mu$ m) are present in quartz (Kotov et al., 2015b).

- 4. Geochemical studies showed that the behaviour of such elements as Nb-Ta and Zr-Hf, which are parts of the ore minerals, mainly pyrochlore and zircon, has been controlled by magmatic processes. The well expressed linear positive Nb-Ta and Zr-Hf correlations are particularly indicative for that (Fig. 9a and b). Similar Nb/Ta = 10–16 and Zr/Hf = 24–34 ratios in the studied granites regardless of absolute concentrations of these elements are also in favour of the dominance of magmatic processes. In addition, the absence of correlation between SiO<sub>2</sub> and Nb, Zr, as well as between Na<sub>2</sub>O and Nb, Zr (Fig. 9c–f), suggest very little influence of the postmagmatic (metasomatic) processes (e.g. greisenization, albitization) on the behaviour and distribution of these elements (Huang et al., 2014).
- 5. The 2066  $\pm$  6 Ma U-Pb zircon age of the Aeg-Arf granites (Larin et al., 2002) is close to the 2055  $\pm$  7 Ma zircon age of the ore segregation, suggesting that the formation of the Katugin rare-metal deposit is genetically related to the forming processes of the per-alkaline granites.

## 8.2. A model proposed for the rare-metal Ta-Nb-Zr-REE mineralization of the Katugin deposit

Fluorine is the important factor (even a key factor) of the rare-metal (Ta-Nb-Zr-REE) enrichment of the acid melts, because the fluorine is a complexing agent for these elements, which is able to increase significantly their solubility (e.g., Manning, 1981; Charoy and Raimbault, 1994; Webster et al., 2004; Agangi et al., 2010; Moghazi et al., 2011). The Arf, Aeg-Arf and Aeg granites show high concentrations of F both in the rocks (up to 3.30 wt%) and in rock-forming and accessory minerals: arfvedsonite (up to 2.55 wt%), pyrochlore (up to 5.50 wt%) and bastnaesite (up to 8.61 wt%) This is likely related to dissolution of the fluorine in the magmatic melt at all stages of the magmatic evolution, including the late magmatic stage during the crystallization of the

majority of ore mineral phases enriched by fluorine. Besides, some negative correlation between SiO<sub>2</sub> and F (r = -0.72) and consequently a small amount of fluorine in mostly felsic rocks indicates that the increasing of fluorine content in rocks is not related to post-magmatic hydrothermal processes. If the fluorine is added to a high temperature melt, it dissolves in it causing a decrease of the solidus temperature and changes in physical properties of acid magmas, in particular - a decrease of viscosity and an increase of diffusivity (Manning, 1981; Webster, 1990; Xiong et al., 1999; Scaillet and Macdonald, 2001; Lukkari and Holtz, 2007; Huang et al., 2014). Consequently a decrease of the solidus temperature causes a crystallization of major mineral phases at lower temperatures, which in turn makes Ta-Nb, Zr-Nb, REE incompatible to the melt and causes their assemblage to mineral complexes with the dissolved fluorine (Manning, 1981; Webster et al., 1989, 2004; Markl et al., 2001; Schönenberger et al., 2008; Agangi et al., 2010; Gramenitskiy et al., 2005; Shchekina et al., 2013; Dostal et al., 2014; Dostal and Shellnutt, 2016). At the early stages of magmatic crystallization these mineral complexes preserve their melt texture and are randomly distributed within magmatic system due to a low viscosity and high diffusivity of the F-rich granitic magma. Consequently they cannot crystallize at earliest stages, causing increase of the rare elements in the residual melt (Huang et al., 2014). Therefore, the majority of rare-metal accessory minerals crystallize mostly at the late magmatic stage.

### 8.3. A model for various types of cryolite segregation in the Eastern Katugin massif

The high content of cryolite is specific for the granites in the eastern segment of the Eastern Katugin Massif. This indicates high fluorine content in the source melts. As it was mentioned before, cryolite occurs in granites both as stringer disseminated pockets (accessory cryolie), and as larger veins or lens-like bodies of cryolite-rich rocks (latest magmatic cryolite). The accessory cryolite in granites occurs as inclusions in quartz, fills spaces between grains of leucocratic minerals, develops on the margins of feldspars and fills crevices in early magmatic albite and often in intergrowths with aegirine. Such petrographic data suggest a crystallization of the accessory cryolite from a granite melt enriched by fluorine, where cryolite is in equilibrium with main



Fig. 9. Ta - Nb (a), Hf - Zr (b), Nb - SiO<sub>2</sub> (c), Zr - SiO<sub>2</sub> (d), Nb - Na<sub>2</sub>O (e), Zr - Na<sub>2</sub>O (d) for Arf, Aeg-Arf and Aeg granites of the eastern segment of the Eastern Katugin massif.

rock forming minerals (Shchekina et al., 2013). Good positive correlation between Na<sub>2</sub>O and F in granites (Fig. 3e) suggests their coherent behaviour in the magmatic system. Apart from an accessory cryolite there are some larger veins or lens-like bodies of cryolite-rich rocks in the eastern segment of the eastern Katugin massif, which never extend outside of the massif. One of such bodies up to 10 m thick is reported from the southern part of the eastern segment. This body, apart from cryolite, contains crystals and intergrowths of arfvedsonite, annite, bafertisite, lepidolite, potassium feldspar, quartz, pyrochlore, gagarinite, tveitite, ilmenite, pyrite, sphalerite, magnetite and barite. Geological observations suggest that large cryolite bodies have been formed at the final stages of the formation of Katugin deposit, so we can consider them as largest magmatic bodies. Shchekina et al. (2013) demonstrated that such bodies could be formed during crystallization of cryolite from the fluorine salt melt separated from the aluminosilicate melt at the late stage of its differentiation. The residual melt crystallized

later at lower temperatures than the silicate melt.

### 8.4. Source of the granites from the eastern segment of the Eastern Katugin massif

Peralkaline ferroan composition of the studied Afr, Aeg-Arf and Aeg granites, their isotopic ( $\varepsilon$ Nd(t) = 0.0... – 1.9) and geochemical (Y/Nb = 0.01–1.38, Rb/Nb = 0.17–2.25) characteristics suggest that basaltic magmas of OIB type (possibly with some influence of the crustal contamination) represent a dominant part of the granitic source. Enriched mantle source is supported by principal geochemical ratios, in particular by lower Th/Ta and Zr/Nb (0.1–2.0 and 0.4–8.4 correspondingly), indicating that this source was represented by the OIB-type mafic magmas (Larin et al., 2012). Strongly differentiated granite's compositions (Rb/Sr = 17–640, Rb/Ba = 5–159) indicate that the parent basaltic magmas should be produced at a high degree of the

fractional crystallization to achieve the stage of a granite melt. However the peralkaline silicic melt is not enough for the forming of rare-metal granitoids. The large amount of fluorine in the melt is also required. In spite of the increasing of the fluorine content during a progressive differentiation of the magmatic series of the felsic rocks, it is often not enough for the formation of large rare-metal deposits. We suggest that the fluorine-enriched magmatic fluid phases could provide an additional source of the fluorine. Dissolution of fluorine in the silicate melt has led to crystallization of rare metals and cryolites.

#### 8.5. Tectonic setting of the Katugin rare-metal deposit

It is generally accepted that a great majority of large rare-metal deposits in the Earth are related to the alkaline rocks such as carbonatites, nepheline syenites and alkaline granites (see overview in Pirajno, 2009). According to modern ideas, either mantle plume, or sub-continental lithospheric mantle enriched in incompatible elements as the result of the plume-related activity, or of oceanic subduction, can be the sources for these rocks (Simonetti et al., 1998; Graham et al., 2004; Castor, 2008; Smith et al., 2016).

To decipher the geodynamic environments of the formation of the Katugin granites and of the Ta-Nb-Y-Zr (REE) deposit in particular, we have to consider the following: 1) analysis of the geological activity of the area in time of the granite emplacement, and 2) pertrological, geochemical and isotopic features of the granites.

The analysis of the evolution of superterranes in the southern Siberia and the distribution of major peaks in magmatic and tectonic activities in these superterranes (Gladkochub et al., 2009) demonstrate that the time around 2.06 Ga (the age of Katugin granites) belongs to the stagnant period of the anorogenic geodynamic environment. Apart from Katugin granites, the only magmatic rocks with ages between 2.20 and 2.06 Ga in the studied area are few picritic dykes (2204  $\pm$  41 Ma, Sm-Nd mineral - whole rock. Puchtel and Zhuravley, 1992) enriched in HFS elements. Such low magmatic activity on the wide areas of the southern part of the Siberian craton supports the hypothesis about the domination of anorogenic environments there with just minor extensional events (Larin et al., 2002; Kotov, 2003). Importantly the mentioned 2.20-2.06 Ga magmatic events expressed by picrite dykes and Katugin granites were not directly related to the formation of Siberian craton, but resulted from just local internal processes within terranes, which assembled after 1.88-1.85 Ga accretionary and collisional events.

The conclusion about the anorogenic nature of the studied raremetal granites, which is based on geological data and on the likely scenario of the Siberian Craton's formation, is supported by geochemical data. Notably it is possible, that the anorogenic magmatism is often related to mantle plumes (Pirajno, 2015). Our geochemical and isotopic data support plume-related nature of the Katugin Massif. Interestingly there is no evidence for a coeval magmatic activity in other areas of anywhere in the Siberian craton (Gladkochub et al., 2009, 2010). This suggests that the mantle plume was most likely localized beneath the allochthonous terrane which hosts the Katugin deposit and that this terrane at ~2.05 Ga has not been connected to the southern part of the Siberian Craton, but developed and moved separately at some distance.

Felsic and alkaline magmatism was not wide spread at  $\sim 2.05$ Ga, but several large rare-metal deposits are associated with these types of magmatism. Apart from the Katugin deposit there are at least four other nearly coeval deposits. The closest analogue is the Canadian Thor Lake group of deposits. They are located in the southern part of the Archean Slave Craton near its margin within the  $\sim 2.0-1.8$  Ga Thelon orogenic belt (e.g. Whitmeyer and Karlstrom, 2007; Buchan et al., 2010). Five deposits of this group with complex Zr, Nb, Ta, REE, Be and Ga mineralization, which belong to the unique category (Osokin et al., 2000), are associated with the alkaline to peralkaline Blachford Complex, related to a development of the Athapuscow aulacogen (Trueman and Pedersen, 1988). The age of the older part of this complex is

 $2175 \pm 7$  Ma, the age of the younger part –  $2094 \pm 10$  Ma (Bowring et al., 1984). The rare-metal ores, which are associated with alkaline and nepheline syenites, have the similar age (Sheard et al., 2012). The alkaline rocks of the Blachford Complex have been formed in the extensional environments, which are indicated by the 2.03–2.02 Ma mafic dyke swarms (Buchan et al., 2009).

Mount Weld in Western Australia is another giant rare-metal (REE, Nb, Ta, Zr) deposit of similar age. The ore zone is a laterite developed after carbonatite in Quaternary time. The carbonatite intrusion cuts across the Archean volcano-sedimentary rocks of the Laverton tectonic zone and belongs to the Paleoproterozoic magmatic province, which includes alkaline intrusions of carbonatites and kimberlites (Duncan and Willett, 1990). The Re-Os age of the Mount Weld carbonatites is  $2025 \pm 10$  Ma (Graham et al., 2004), and the Rb-Sr age is  $2021 \pm 10$  Ma (Collerson, 1982).

The rare-metal deposits are also associated with the ~2060 Ma giant bimodal Bushveld Complex in South Africa (Kaapvaal Craton). The Lebowa Granite Suite, which is co-genetic with the mafic to ultramafic layered Rustenburg Series, forms a 2–3 km thick plate-like body in the central part of the complex. These alkaline granites are enriched in incompatible elements and belong to the intraplate A-type granites. The commercially exploitable Sn-fluorite, Sn-polymetallic, Mo-fluorite and uranium deposits are associated with these granites, while the layered series of the Bushveld Complex contains the world largest chromium, vanadium and PGE deposits. It is also possible to associate the alkaline-ultramafic intrusions and carbonatites of the Phalaborwa Complex with the Bushveld Complex – they are spatially close and have a similar age of 2047  $\pm$  8 Ma (Eriksson, 1984). The Phalaborwa carbonatites are associated with commercially exploitable Cu, REE, U, Zr, Au and apatite mineralization (Harmer, 2000).

Interestingly the cratons on which the abovementioned  $\sim 2.05$  Ga giant rare-metal deposits are found (i.e. Slave, Yilgarn and Kaapvaal) have been assigned by Bleeker (2003) to three different craton "clans" with little, if any similarities between them. Hence any paleogeographic affinity between these deposits is not very likely. However, the present state of the knowledge about the  $\sim 2.05$  Ga global paleogeography is quite low, so any speculations on this matter are probably premature.

#### 9. Conclusions

- 1. The geological, mineralogical, geochemical and geochronological data are indicative of the magmatic genesis of the giant Katugin rare-metal deposit rather than of previously suggested (Prokhorov and Sobachenko, 1985; Arkhangel'skaya et al., 1993, 1998) unique metasomatic fault-related origin.
- 2. The Ta-Nb-Y-Zr (REE) ores of the Katugin deposit are genetically related with the granites of the Katugin complex, mainly with arf-vedsonite, aegirine-arfvedsonite and aegirine granites.
- 3. The studied Arf, Aeg-Arf and Aeg granites of the eastern segment of the Eastern Katugin massif correspond to peralkaline granite. They are characterized by high alkalinity, high iron content (FeO<sup>\*</sup>/ (FeO<sup>\*</sup> + MgO) = 0.96–1.00), very high content of most incompatible elements Rb, Y, Zr, Hf, Ta, Nb, Th, U, REEs (except for Eu) and F, and low concentrations of CaO, MgO, P<sub>2</sub>O<sub>5</sub>, Ba, and Sr. They demonstrate negative and CHUR-close  $\varepsilon$ Nd(*t*) values of 0.0... 1.9. Some aegirine and aegirine-arfvedsonite granites have extremely high contents of Zr, Hf, Nb, Ta and total REE.
- 4. Basaltic magmas of OIB type (possibly with some influence of the crustal contamination) could be a dominant part of a source for Aeg, Aeg-Arf and Arf granites. Moreover, the fluorine-enriched fluid phases could provide an additional source of the fluorine. Strongly differentiated granite's compositions indicate that the parent basaltic magmas should be produced at a high degree of the fractional crystallyzation.
- 5. The results of detailed mineralogical study suggest three major types of ore-mineralization in the Katugin deposit: Zr mineralization

(zircon), Ta-Nb-REE mineralization (pyrochlore, bastnaesite, columbite, fluocerite, tveitite, gagarinite etc.) and aluminum fluoride mineralization (cryolite etc.).

- 6. Most of ore minerals crystallized from the silicate melt during the magmatic stage. The accessory cryolites in granites have been crystallized from the magmatic silicate melt enriched with fluorine. However, cryolites in large veins and lens-like bodies have been crystallized at the latest stage from the fluorine salt melt.
- 7. The zircons from ore segregation in the Aeg-Arf granite have been dated at 2055  $\pm$  7 Ma. This age is close to the 2066  $\pm$  6 Ma zircon age of the granites themselves (Larin et al., 2002), suggesting that the formation of the Katugin rare-metal deposit is genetically related to the forming processes of the peralkaline granites.
- The geological and geochemical data prove an anorogenic nature of the Katugin rare-metal granites related to a Paleoproterozoic (~2.05 Ga) mantle plume occurred in the studied area.

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#### Appendix A. Supplementary data

Supplementary data associated with this article can be found, in the online version, at http://dx.doi.org/10.1016/j.oregeorev.2017.10.002.

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