# Rare earth elements in rocks and minerals from alkaline plutons of the Kola Peninsula, NW Russia, as indicators of alkaline magma evolution

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Abstract. In order to elucidate evolutionary paths for the alkaline ultramafic series of the Kola province, we studied distribution of rare earth elements (REE) in rocks and constituent minerals of the rock sequence dunite, clinopyroxenite, melilitolite, meltejgite, ijolite, nepheline svenite. Abundances of REE and other trace elements were measured in olivine, melilite, clinopyroxene, nepheline, apatite, perovskite, titanite, and magnetite. Distribution of most trace elements in Kovdor-type rocks is shown to differ fundamentally from that in the Khibiny alkaline ultramafic suite and to have been controlled by perovskite crystallization. Primary olivine melanephelinitic melts of the Kovdor series are demonstrated to be characterized by early crystallization of perovskite, the most important REE mineral. Perovskite co-precipitating with olivine and clinopyroxene leads to a sharp REE depletion of the residual melt, to produce REE-depleted derivatives, ijolites and nepheline syenites. By contrast, the genesis of the Khibiny alkaline ultramafic series was complicated by mixing of minor batches of phonolitic melt with the primary olivine melanephelinitic magma, which led to changes in the crystallization order of REE-bearing titanates and Ti-silicates and to enrichment of late melt batches in the most incompatible elements. As a result, Khibiny ijolites have the highest REE abundances, which are accommodated by high-REE apatite and titanite.

# 1. Introduction

In the northeastern Baltic Shield, the Paleozoic stage of tectonic and magmatic reactivation involved generation of alkaline plutonic complexes, which have been customarily divided into two rock suites: (i) alkaline ultramafic rocks associated with carbonatites (Kovdor, Vuoriyarvi, Afrikanda, Seblyavr, etc., massifs) and (ii) agpaitic nepheline syenites, represented in the vast Khibiny and Lovozero plutons. Available isotope ages indicate that all the Paleozoic alkaline massifs of the Kola province were coeval, and their parental melts were derived from the same mantle sources [Kramm and

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Kogarko, 1994; Kramm et al., 1993]. The hallmark of the region's alkaline rocks is their immense abundances of REE, Y, Sr, Zr, Hf, Nb, Ta, and Th. These are either concentrated in apatite, titanite, perovskite, and other accessories or, given their concentrations in melts were high enough, form their own discrete minerals, such as loparite, pyrochlore, and eudialyte, whose economic deposits provide the basis for the region's mining industry.

Geologic observations and experimental evidence suggest that alkaline ultramafic rocks forming parts of carbonatite complexes originated through crystal fractionation in nephelinitic melts, which resulted, at early stages, in olivine, clinopyroxene, and melilite cumulates and their complementary foidolites and nepheline syenites [Arzamastsev et al., 2001; Dawson et al., 1995; Ivanikov et al., 1998; Kukharenko et al., 1965; Nielsen, 1994; Verhulst et al., 2000]. Available data on behavior of incompatible elements during magmagenesis suggest that with advancing magma crystallization, elements such as Sr, Zr, Hf, Nb, Ta, Th, and REE are likely to become enriched in terminal melt derivatives of the



Figure 1. Location scheme for Paleozoic igneous occurrences in northeastern Baltic Shield. Intrusions: 1 – Khibiny, 2 – Lovozero, 3 – Niva, 4 – Mavraguba, 5 – Kovdor, 6 – Sokli, 7 – Sallanlatva, 8 – Vuoriyarvi, 9 – Kandaguba, 10 – Afrikanda, 11 – Ozernaya Varaka, 12 – Lesnaya Varaka, 13 – Salmagora, 14 – Ingozero, 15 – Cape Turiy, 16 – Kurga, 17 – Kontozero, 18 – Ivanovka, 19 – Seblyavr. Terrane location scheme, after [*Balagansky et al.*, 1998].

alkaline ultramafic series. This is indeed the case in alkaline ultramafics of the Khibiny massif, at whose lower horizons large fragments of such bodies are found surrounded by agpaitic syenites [Arzamastsev et al., 1998; Galakhov, 1975]. However, in the majority of alkaline ultramafic plutons region-wide, terminal crystallization products (ijolites, nepheline syenites, and cancrinite syenites) are appreciably depleted in elements such as Nb, Ta, and rare earths.

This work is intended to study how the above trace elements behave in alkaline ultramafic suites and, in particular, to find out why rare earth elements follow different distribution patterns in ultramafic carbonatite intrusions and in the Khibiny massif. Our study draws on mineralogic and geochemical data on representative samples from the Kovdor, Vuoriyarvi, Cape Turiy, Salmagora, and Afrikanda massifs, the Lesnaya Varaka and Ozernaya Varaka intrusions, and the Khibiny complex. ICP-MS analyses were made on wholerock samples and on separates of coexisting perovskite, apatite, titanite, clinopyroxene, melilite, olivine, nepheline, and magnetite. Data obtained enable us to identify those factors responsible for the diversity of REE distribution patterns in alkaline ultramafic suites and, in particular, to assess the role of high-REE accessory phases–loparite, apatite, and titanite.

## 2. Geologic Structure of the Intrusions

In the northeastern Baltic Shield, intracratonic magmatism that spanned some 40 to 50 m.y. can be divided into three episodes. The initial episode, from 405 to 380 Ma, coinciding with the final phase of Caledonian orogeny, involved inception of the Khibiny, Lovozero, and Kontozero calderas in the foreland of the Caledonian front, accompanied by subalkaline volcanism and emplacement of ultramafite and Nesyenite intrusions. The principal period of igneous activity, between 380 and 360 Ma, gave rise to the multiphase Khibiny and Lovozero plutons and to alkaline ultramafic intrusions associated with carbonatites [Kramm and Kogarko, 1994; Kramm et al., 1993] (Figure 1). The final episode resulted in dike swarms and diatremes made up of alkaline picrites,



**Figure 2.** Scheme showing geologic structure of the Kovdor massif, simplified after [*Ternovoi*, 1977]. 1 – carbonatite, 2 – phoscorite, 3 – nepheline syenite, 4 – foidolite, 5 – olivine–diopside–phlogopite rocks, 6 – olivine–monticellite–melilite rocks, 7 – melilite rocks, 8 – pyroxenite, 9 – peridotite and dunite, 10 – Precambrian basement.

melanephelinites, melilitites, and kimberlites [Arzamastsev et al., 2001].

## 2.1. Alkaline Ultramafic Plutonic Suites Forming Parts of Carbonatite intrusions (Kovdor Type)

Plutonic alkaline ultramafic suites are found forming parts of multiphase intrusions. Lithologic zonation of the intrusive bodies portrays the following order of emplacement of rocks that constitute the series (i) dunite, (ii) pyroxenite, (iii) melilite-bearing rocks (turjaite, melilitolite, okaite), (iv) melteigite, (v) ijolite, (vi) nepheline/cancrinite syenite, and (vii) carbonatites and phoscorites. The complete set of lithologies is represented in the Kovdor, Cape Turiy, and Vuoriyarvi massifs, whereas in the rest of intrusions only particular groups of rock varieties are exposed at the present topographic surface. In most intrusions, olivine and clinopyroxene cumulates make up their cores, melilite rocks and foidolites occurring in peripheral zones (Figure 2). Nepheline and cancrinite syenites either make up detached satellite bodies near the intrusions (e.g., the Maly Kovdor) or fill in veins in alkaline ultramafics (Ozernaya Varaka). Boss-like carbonatite bodies, surrounded by carbonatite stockworks, usually are located centrally in the intrusions, but occasionally they are displaced from geometric centers of the ring bodies of alkaline ultramafites. The order of emplacement, based on generalized field evidence from all the intrusions of the region, is consistent with that reported for alkaline ultramafic intrusions worldwide [Kogarko et al., 1995; Le Bas, 1987; Nielsen, 1987; Woolley, 1987]. Alkaline ultramafic intrusions occur in spatial association with bodies of olivine melteigite porphyry and with dikes and diatremes of alkaline picrite, Ol-melanephelinite, nephelinite, and melilitite.

## 2.2. Alkaline Ultramafic Plutonic Suites Forming Parts of Agpaitic Syenite Complexes (Khibiny Type)

Although exposed portions of the Khibiny and Lovozero plutons are dominated by agpaitic syenites, geophysical data verified by drilling reveal alkaline ultramafic rocks [Galakhov, 1975] and carbonatites [Dudkin et al., 1984] within the Khibiny massif and ultramafic and melilite rocks [Arzamastsev et al., 1998] within the Lovozero massif. According to our own data [Arzamastsev et al., 1998], volumetrically, alkaline ultramafic rocks make up at least 30% of the Khibiny pluton and 25% of the Lovozero pluton within the 12.5 km depth range accessible to gravity surveying. The plutons are thus shown to comprise the complete series of rocks typical of alkaline ultramafic massifs of the province: peridotites, pyroxenites, melilitolites, melteigites, ijolites, and carbonatites. The Khibiny complex displays at least three phases of emplacement of alkaline ultramafic melts with intervening stages of Ne-syenite magma injection (Figure 3). The peridotites, pyroxenites, and melilitolites originated at the earliest formative phase of the massif (Phase I, Figure 3b), which preceded agpaitic syenite injections. The ring melteigitesijolite intrusion (Phase II, Figure 3d) took shape after the emplacement of nepheline syenites forming the margin of the Khibiny massif, but prior to the formation of the Ne-syenites found in its core. The Khibiny alkaline ultramafic series culminated in carbonatites (Phase III, Figure 3g), which cut through the Ne-syenite core of the massif and carry pyroxenite, ijolite, and melteigite xenoliths.

As regards the Lovozero massif, alkaline ultramafics are encountered mostly in its northeast part, where large blocks and xenoliths of peridotites and melilitic rocks occurring among nepheline syenites suggest that these rocks originated at the earliest stage of emplacement of the complex. Foidolites are not widespread in the massif, their presence being evidenced by sporadic, Ne-syenite-hosted, ijolite and melteigite xenoliths, recovered in cores from a number of drillholes.

# 3. Analytical Techniques and Sample Preparation

Sixty-nine representative samples, collected from drillcores and outcrops in the Khibiny, Lovozero, Kovdor,



Figure 3. Hypothetical model for the formation of the Khibiny complex [Arzamastsev et al., 1998]. Symbols: 1 – carbonatite, 2 – pulaskite, 3 – nepheline syenite in the core of the massif (foyaite), 4 – kalsilite/nepheline syenite, juvite, and apatite rocks, 5 – ijolite and melteigite, 6 – nepheline syenite in the rim of the massif, 7 – olivine pyroxenite and melilitolite, 8 – alkaline volcanites, 9 – Precambrian basement, 10 – fault. (a)–(g) – phases of emplacement of the massif.

Afrikanda, Cape Turiy, Lesnaya Varaka, Ozernaya Varaka, Seblyavr, Salmagora, Sallanlatva, Vuoriyarvi, and Ivanovka intrusions, were selected for this study from a total of more than 500 samples. Modal-mineral compositions of the samples are listed in Table 1.

Whole-rock major element analyses were carried out

at the Geological Institute, Kola Science Center, Russian Academy of Sciences, using a routine technique of sample fusion with  $2Na_2CO_3+1Na$ -tethraborate followed by dissolution in HCl. Si, Al, Mg, Ca, Fe, Ti, Ni, Co, Cr, and V were measured by atomic absorption on a Perkin Elmer 403 instrument with an accuracy (coefficient of variation, CV) better than 3% (G. Gulyuta, analyst). Na, K, Li, Rb, and Cs were measured by flame photometry (CV ~5%); P and S, by photocolorimetry and polarography from the same solutions with an accuracy of ~5% and ~10%, respectively; and F and Cl, by the ion-selective electrodes method (CV ~15%). The Fe<sup>2+</sup>/Fe<sup>3+</sup> ratio was measured by titration (CV ~10%). H<sub>2</sub>O and CO<sub>2</sub> were analyzed by the gravimetric method (CV ~10%).

Trace elements were analyzed by ICP-MS at the University of Granada. Sample charges of 0.1 g were allowed to stand for 30 min in an HNO<sub>3</sub> + HF mixture in Teflon-lined containers at T =  $180^{\circ}$ C and  $\sim 200$  p.s.i., evaporated until dry, and dissolved in 100 ml of 4% HNO<sub>3</sub>. Each specimen was analyzed three times on an ELAN-5000 PE SCIEX instrument using a rhenium within-lab standard. Accuracies were  $\pm 2$  rel.% and  $\pm 5$  rel.% or better for concentrations of 50 and 5 ppm, respectively.

Major elements in minerals were measured at the Geological Institute, Kola Science Center, on a Cameca MS-46 ion microprobe using natural and synthetic standards. Acceleration voltage was 30 kV for Sr and Zr and 20 kV for the rest of the elements; sample current was 20–40 nA, and ion beam diameter, 1.5–3  $\mu$ m.

Mineral grains separated for trace-element analysis were inspected under an optical microscope in order to reject foreign inclusions. A number of samples were inspected under a scanning electron microscope. Minerals were separated on a magnetic separator and in heavy liquids. Finally, 8–10 mg charges were cleaned by repeated hand-picking to 99.9 vol % purity. Trace element abundances in sample charges were measured by ICP-MS as described above.

# 4. Petrography of the Rocks and Distribution of REE-bearing Mineral Phases

Ultramafic rocks. Kovdor-type dunites, Ol-clinopyroxenites, and pyroxenites are adcumulates and mesocumulates with olivine and clinopyroxene as cumulus phases and ore minerals, phlogopite, spinel, and nepheline as intercumulus (Table 1). The only primary high-REE minerals are perovskite and apatite. In the dunites, perovskite occurs sporadically as an intercumulus phase forming small roundish grains (Figure 4a). In the pyroxenites, perovskite is present as an early cumulus phase accounting for as much as 40 vol % in Vuorivarvi rocks, 19–31 vol % in Afrikanda, and 11-16 vol % in the Salmagora massif [Korobeinikov et al., 1998; Kukharenko et al., 1965]. In ultramafic rocks, apatite is less common, dunites containing no more than 0.2% and pyroxenites usually bearing up to 3% apatite by volume. The only exception is Vuoriyarvi and Afrikanda

Table 1. Modal-mineral compositions of alkaline ultramafic rocks (vol %)

Ol Cpx Mel Ne Mic Prv Ap Ttn Mag	Others
4/400 KVD Dunite 92 <1 8	Spl 1
227/43 KVD Turjaite 5 57 5 20 1 <1 12	~F
252/100 KVD Turjajte 3 56 3 33 <1 <1 3	
7/394 KVD Foidolite 65 20 8 <1 7	
7/348 KVD Foidolite 65 10 10 <1 2 3 4	Amph 5
5/740 KVD Foidolite 77 5 7 3 3 5	<b>r</b> -
5/610 KVD Foidolite $37$ $55$ $3$ $2$ $3$	
7/30 KVD Foidolite 58 35 2 1	Cal 5
7/92.8 KVD Foidolite 69 20 1 <1 5 3 1	0
27/65 KVD Foidolite 5 70 10 5 <1 4 5	
MK-10 KVD Nepheline svenite 20 35 10 1 3 1	Fsp 30
AFR-1 AFR Peridotite	1
AFR-5 AFR Pyroxenite 70 8 17 1 5	
AFR-6 AFR Pyroxenite 70 14 5 3	Grt 8
25-AFR AFR Pyroxenite $90$ $2$ $5$ $1$ $1$	Cal 1
Z-6 LSV Dunite 95 <1 5	
SL-23 SLM Dunite 87 1 1 <1 10	Spl 1
SL-24 SLM Pyroxenite 75 5 5 3 5 1 6	~F
93/192 VUO Pyroxenite 70 2 10 <1 18	
$\frac{282}{224}$ VUO Pyroxenite 84 12 1 10	
282/251 VUO Pyroxenite 66 5 10 1 18	
257/208 VUO Melteigite 65 30 2 1 1	
25/15 SLN light 50 44 1 2 2	Cal 1
26/151 SLN Ijolite $40$ 50 1 1 9	
31/200 SLN lighter 45 40 2 1 2 5	Cal 5
26/151 SLN lighte $37$ 52 1 3 2 4	Cal 1
8-OV OZV Melteigite $71$ 5 5 <1 8 5 5	0
1010/1052 KHI Peridotite 60 20 $10 < 1$ 1 6	
A-1044 KHI Peridotite $40$ $24$ $5$ $29$ $1$ $1$	
A-1036 KHI Pyroxenite 57 3 15 <1 1 5	Amph 18
A-1087 KHI Pyroxenite 67 5 15 6 2 5	P
A-1038 KHI Pyroxenite 3 39 2 30 1 1 10	Amph 14
1010/1059 KHI Turiaite $5$ 50 $20$ <1 2 $25$	Grt 1
1010/1165 KHI Turjaite $5$ 45 20 <1 <1 15	Mtc 10, Spl 3
455/345 KHI Melteigite 65 23 2 3 7	· · · · · · · · · · ·
466/552 KHI Melteigite $60$ 18 10 2 5 5	Fsp < 1
1119/279 KHI Melteigite 66 11 5 7 8	Fsp 3
1010/1186 KHI Ijolite 24 20 30 1 3 3 19	Fsp 3
1072/530 KHI liolite 50 36 1 9 2	Fsp 2
455/402 KHI Ijolite $45$ $45$ $3$ $1$ $3$ $1$	Fsp 2
1636/721 KHI Ijolite $21$ $55$ $2$ $3$ $3$ $2$	Fsp 14
1152/83 KHI Ijolite 33 47 1 1 5 3	Fsp 10

Note: Mineral composition were determined by thin-section counting on 1000 points. Hereinafter in Tables: KVD – Kovdor, LSV – Lesnaya Varaka, SLM – Salmagora, AFR – Afrikanda, VUO – Vuoriyarvi, OZV – Ozernaya Varaka, SLN – Sallanlatva, KHI – Khibiny. Ol – olivine, Cpx – clinopyroxene, Mel – melilite, Mic – mica, Ne – nepheline, Prv – perovskite, Ap – apatite, Mag – magnetite, Mtc – monticellite, Spl – spinel, Ttn – titanite, Grt – Ca-Ti-garnet, Cal – calcite, Fsp – K-Na-feldspar, Amph – amphibole.

pyroxenites, in which apatite accounts for as much as 8% of rock volume [Kukharenko et al., 1965]. Besides perovskite and apatite, the pyroxenites contain sporadic titanite that forms secondary segregations partially replacing perovskite and magnetite.

Ultramafic rocks encountered in the Khibiny and Lovozero

massifs are represented by peridotites and, chiefly, by pyroxenites, which are texturally and compositionally similar to ultramafics of the Kovdor suite. Pyroxene with characteristic cumulus features is in places partially replaced by richterite and/or phlogopite. REE-bearing accessories are represented by apatite and titanite (up to 6 and 2 vol %, ARZAMASTSEV ET AL.: RARE EARTH ELEMENTS IN ROCKS AND MINERALS



**Figure 4.** Perovskite segregations in rocks of the alkaline ultramafic series (backscattered electrons raster photomicrographs): a – primary perovskite (Prv) in olivine (Ol) from Kovdor dunite, sample 4/400; b – secondary perovskite overgrown on magnetite (Mag) and melilitolite + phlogopite (Phl) grains, Kovdor, sample 252/100.

respectively; Table 1). An important distinction is that perovskite is a rare accessory in all ultramafic rocks from the Khibiny.

Melilite-bearing rocks, dominantly uncompahyrites and turjaites, form independent intrusion phases in the Kovdor and other alkaline ultramafic massifs. In the Khibiny and Lovozero, melilite rocks are encountered as xenoliths. Melilite together with phlogopite compose large oikocrysts with clinopyroxene and nepheline inclusions. REE-bearing phases, represented by perovskite, apatite, and, less frequently, titanite, occur sporadically. Intersticial segregations of magnetite are not infrequently rimmed by secondary perovskite (Figure 4b), indicative of a Ti-bearing phase having been exsolved from primary titanomagnetite [Nielsen et al., 1997].

Foidolites occur widely in both the Kovdor and Khibiny alkaline ultramafic suites. The rocks are ortho- and mesocumulates with nepheline segregations varying in habit from anhedral to euhedral in less and more leucocratic rocks, respectively. Clinopyroxene makes up zoned segregations whose compositions are different in the Kovdor and Khibiny suites. In foidolites of the Kovdor and other alkaline ultramafic massifs, pyroxene is represented by diopside, which makes up grain cores, while grain rims are composed of aegirine-augite. Pyroxenes in Khibiny ijolite-melteigites are more alkaline than in the Kovdor suite and, unlike the latter, are composed of aegirine-augite rimmed by aegirine. Accordingly, late amphiboles developed after clinopyroxene have different compositions, mostly pargasitic in Kovdor foidolites and corresponding to richterite or magnesiocataphorite in the Khibiny suite.

Distribution patterns of REE-bearing accessories are different in Kovdor and Khibiny foidolites. In Kovdor, Vuoriyarvi, and Cape Turiy melteigites and ijolites, the earlygeneration perovskite, just as apatite and titanite, is a typical accessory phase. Not infrequently, the perovskite is observed to be replaced by titanite. The mean titanite abundance in ijolites is <1 vol %. Apatite abundances in foidolites attain a critical maximum of 1.2 wt % in the most melanocratic lithologies [Arzamastseva and Arzamastsev, 1996].

Unlike the Kovdor suite, Khibiny foidolites contain apatite and titanite not only as late magmatic accessories, but also as widespread primary REE minerals. Both apatite and titanite are most abundant in melanocratic foidolite varieties, where they account for 4 and 5 vol % on average, respectively. A distinctive feature of Khibiny foidolites is that they lack primary perovskite.

Nepheline- and cancrinite syenites in Kovdor-type massifs show evidence of early crystallization of light-colored phases, nepheline and K–Na-feldspar. Just as in foidolites, pyroxene is represented by zoned grains of diopside rimmed by aegirine-augite. REE-bearing accessories are titanite and apatite.

Arzamastsev et al. [1998] showed that neither Khibiny nor Lovozero agpaitic Ne-syenites are cogenetic with rocks of the alkaline ultramafic series. This is evidenced, in particular, by the fact that the REE-bearing mineral assemblage in agpaitic syenites differs fundamentally from that in the alkaline ultramafic rocks. Lovozero lujavrites have as much as 90 vol % eudialyte and 12 vol % loparite, while in Khibiny Ne-syenites the most widespread primary magmatic phases are eudialyte and apatite. Distribution, composition, and origin of these economically important minerals were studied by Kogarko [1977, 1999] and Kravchenko et al. [1992] and are beyond the scope of this work.

## 5. Results

#### 5.1. Rock Chemistry

**5.1.1. Major elements.** Bulk-rock analyses representative of Kovdor- and Khibiny-type alkaline ultramafic suites are listed in Tables 2 and 3. Compositional variation trends for both suites, plotted on the totality of major-element analyses amassed to date (Figure 5), show the ultramafic portion to be controlled by precipitation of olivine and clinopyroxene, whereas the foidolite trend is controlled by fractionation of clinopyroxene and nepheline. The decrease in Mg# through this rock sequence from 0.90 in dunites to 0.56 in Ne-syenites of the Kovdor suite is correlative with variations of Ni, Cr, Co, V, and Sc.

The following distinctions exist between the Kovdor and Khibiny suites in terms of major-element abundances. Khibiny foidolites are relatively lower in MgO and higher in  $SiO_2$  and alkalis (Figure 5). Thus, weighted mean abundances of these oxides in Khibiny foidolites are 43.62 wt % $SiO_2$ , 9.40 wt % Na<sub>2</sub>O, 3.59 wt % K<sub>2</sub>O, and in Kovdor-type foidolite intrusions, 41.64 wt % SiO<sub>2</sub>, 8.10 wt % Na<sub>2</sub>O, 2.73wt % K<sub>2</sub>O. The high contents of silica and alkalis are manifest in modal-mineral compositions of Khibiny foidolites, such that K-Na-feldspar is a characteristic accessory mineral in ijolites, where it accounts for as much as 10 vol %. Among other distinctions between Kovdor and Khibiny rocks, one should note the higher F content of Khibiny rocks and the differences in TiO<sub>2</sub> and P<sub>2</sub>O<sub>5</sub> distributions. Thus, in Kovdor suite the highest TiO<sub>2</sub> abundances are detected in pyroxenites, which have 8–15 wt % MgO on average, whereas in Khibiny suite  $TiO_2$  is highest in the most evolved rocks, ijolites and melteigites, which have 3–7 wt % MgO.

**5.1.2.** Rare earth elements (REE). Chondritenormalized plots for REE from both rock suites, listed in Tables 4 and 5, are shown in Figure 6. Along with data for specific samples, all the plots display the REE pattern for the weighted mean composition of alkaline ultramafic rocks of the Kola province [*Arzamastsev et al.*, 2001].

All the Kovdor-type alkaline ultramafic rocks lack Eu anomaly and show depletion in the light REE relative to the heavy REE (Figure 6). The lowest total REE contents and  $(La/Yb)_N$  ratios (11.7–17.4) were detected in olivine cumulates from the Kovdor, Lesnaya Varaka, and Salmagora massifs. By contrast, pyroxenites from the majority of massifs, which contain perovskite and, to a lesser extent, apatite, are sharply REE-enriched relative to the mean composition of alkaline ultramafic rocks. These rocks have steeper REE patterns with  $(La/Yb)_N = 52-226$ . More evolved derivatives (melilitolites, foidolites, and Ne/cancrinite symptometers) have lower REE contents compared to the mean alkaline ultramafic rock composition for the province, at  $5 \times 10$  to  $3 \times 10^2$  times the chondritic level. Therefore, with advancing differentiation of the Kovdor suite, late-stage ijolite and Nesyenite derivatives become progressively depleted in REE.

Khibiny alkaline ultramafic rocks have a REE distribution pattern which is sharply dissimilar to the Kovdor suite (Figure 6). By and large, REE abundances of perovskite-

free peridotites and pyroxenites are close to the estimated mean values for the alkaline ultramafic series. Due to low  $(La/Yb)_N$  ratios, ranging 31.4–53.4, the light REE contents are 0.5–0.7 times the mean values for the alkaline ultramafic series, whereas the medium and heavy REE contents are 2 times higher than the mean values. Just as in pyroxenites, REE abundances in Khibiny melilite rocks are within the range of mean values established for the alkaline ultramafic rocks. On the other hand, the latest derivatives in the suite, ijolites and melteigites, are markedly REE-enriched relative to the mean composition of alkaline ultramafic rocks of the province. In particular, melteigites from the Khibiny layered complex, which have up to 8 vol % titanite and 5 vol %apatite, show  $2 \times 10^3$  times the chondritic REE concentrations. To sum up, the Khibiny alkaline ultramafic suite is characterized by a progressive REE enrichment of its late derivatives.

### 5.2. REE distribution in minerals

5.2.1. REE-bearing phases. Perovskite. Two generations of perovskite have been reported from alkaline ultramafic rocks of the province. Generation I perovskites originate from the early igneous stage of alkaline ultramafite crystallization, as evidenced by the fact that melt inclusions in these perovskites match trapped portions of the primary melt compositionally [Kogarko et al., 1991; Veksler et al., 1998b]. Generation II perovskite, found mainly in melilitolites and foidolites, makes up rims around perovskite I and around magnetite. According to [Mitchell, 1996b], the early-generation perovskites are compositionally close to the ideal formula CaTiO<sub>3</sub>, whereas late perovskite segregations follow a loparite trend associated with enrichment in Na, LREE, Nb, and Th. Established are appreciable compositional distinctions between the early and late generations of the perovskites extracted for analysis. Generation I perovskites (Table 6) have the lowest REE, Nb, Ta, Y, U, Th, and Sr abundances, which have been detected in Vuorivarvi pyroxenites. On the other hand, in Generation II perovskites, as represented by samples from Kovdor melilitolites and Ozernaya Varaka ijolites, the listed elements have 2 to 10 times the concentrations of the early magmatic perovskites. Overall, all the perovskites are sharply LREE enriched ((La/Yb)<sub>N</sub> = 207-518) (Figure 7). Compared to perovskites from other carbonatite assemblages worldwide, the varieties under study are closely similar to those from the Oldoinyo Lengai foidolites [Dawson et al., 1995] and kimberlites [Mitchell and Reed, 1988] in terms of REE contents, but the latter have higher  $(La/Yb)_N$  ratios.

Apatite. Besides Ca, P, and F, apatites from all the rocks under study have appreciable abundances of Sr, Y, and REE (Table 6), the late apatite generation having highest values. Apatites from pyroxenites and foidolites of the Kovdor, Afrikanda, Sallanlatva, and Khibiny plutons have very narrow ranges of both total REE contents and  $(La/Yb)_N$ , 57– 278. Chondrite-normalized REE spectra (Figure 7) for apatites are nearly parallel and broadly correlative to the REE abundances of host rocks. The earlier report on Eu anomaly



**Figure 5.** Major oxides (wt %) vs. MgO in alkaline ultramafic rocks of the Kovdor (empty circles) and Khibiny (solid circles) type.

Table 2. Whole-rock major element analyses (wt %) for carbonatite-bearing alkaline ultramafic massifs of the Kola province

Rock	DUN	DUN	DUN	PRD	$\mathbf{PRX}$	PRX	$\mathbf{PRX}$	PRX	PRX	PRX	MELT	MELT	MLG	MLG	MLG
Massif	KVD	LSV	SLM	AFR	AFR	AFR	VUO	VUO	VUO	SLM	KVD	KVD	KVD	KVD	KVD
Sample	4/400	Z-6	SL-23	AFR-1	AFR-5	AFR-6	93/192	282/224	282/251	SL-24	227/43	252/100	7/394	7/348	5/740
SiOa	37 81	38 74	38 58	13.88	37.68	32 17	43 21	34 24	26.97	39.33	41 25	41 10	44 10	48 24	38 35
TiO <sub>2</sub>	0.24	0.35	0.52	22.30	1.69	7.19	4.14	12.28	8.63	3.86	0.79	0.47	1.26	0.22	1.29
Al2O3	1.20	0.38	0.96	0.41	3.36	7.41	3.88	3.63	4.02	7.05	10.90	8.02	10.15	1.24	4.01
Fe <sub>2</sub> O <sub>3</sub>	1.48	3.74	4.90	18.59	4.48	10.95	7.15	8.85	19.09	6.90	3.74	2.68	4.37	2.46	8.78
FeO	9.01	11.22	6.07	11.10	4.81	9.41	4.36	5.76	12.71	9.12	4.38	3.22	6.19	3.43	12.36
MnO	0.20	0.23	0.24	0.23	0.13	0.22	0.14	0.18	0.34	0.21	0.10	0.09	0.15	0.17	0.21
MgO	44.05	43.30	39.67	15.72	8.81	9.18	12.60	11.07	9.69	8.41	10.10	9.90	10.27	14.76	14.41
CaO	4.17	0.69	3.38	14.18	28.87	15.95	22.48	21.56	15.70	20.98	19.60	26.40	14.47	22.50	14.68
Na <sub>2</sub> O	0.10	0.03	0.08	0.09	0.82	1.55	0.50	0.43	0.37	1.68	4.65	3.78	4.56	1.43	0.46
$K_2O$	0.05	0.03	0.51	0.06	0.06	1.29	0.21	0.52	0.52	0.37	2.10	1.35	2.54	0.77	2.02
$P_2O_5$	0.03	0.04	0.04	0.01	8.05	1.01	0.19	0.19	0.16	0.38	0.05	0.07	0.12	1.76	0.08
$CO_2$	0.24	0.16	0.10	0.79	0.20	0.27	0.36	0.26	0.28	0.30	0.28	0.40	0.21	1.73	0.68
S <sub>tot</sub> .	0.08	0.02	0.05	n.a.	0.21	0.25	n.a.	n.a.	0.02	n.a.	n.a.	n.a.	n.a.	n.a.	0.01
F	0.01	0.01	0.01	n.a.	0.40	0.07	0.03	0.04	0.03	0.02	0.06	0.05	0.07	0.18	0.07
$H_2O$	0.72	0.50	3.96	1.82	0.47	3.07	0.44	0.56	0.55	1.33	1.42	1.86	0.88	0.89	2.18
Total	99.39	99.44	99.07	99.18	100.04	99.99	99.69	99.57	99.08	99.94	99.42	99.39	99.34	99.78	99.59
mg#	0.90	0.87	0.92	0.37	0.77	0.64	0.84	0.59	0.36	0.62	0.80	0.85	0.75	0.88	0.68
Rock	M	LG	IJI		IJL	TIL.	IJL	IJ	L LI	L	IJL	NFS	NFS	AV	/ER*
		-				1011		101	101				111.0		
Massif	0	ZV	SLN	. S	SLN	SLN	SLN	KVI	D KVI	) K	VD	OZV	KVD		
Massif Sample	O e 8-0	ZV DV	SLN 31/200	25/	SLN 110	SLN 25/15	SLN 26/151	KVI 5/61	D KVI 0 7/3	ОК 027	VD 7/65	OZV Z-12	KVD MK-10		
Massif Sample SiO <sub>2</sub>	0 e 8-0 45	ZV DV .40	SLN 31/200 41.49	25/	SLN 110 5.93	SLN 25/15 41.93	SLN 26/151 41.25	KVI 5/61 44.5	$\begin{array}{c c} \hline     0 & KVI \\ \hline     0 & 7/3 \\ \hline     6 & 44.4 \\ \end{array}$	O K 0 27 4 4:	VD 7/65 2.92	OZV Z-12 55.79	KVD MK-10 51.83		37.20
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub>	0 e 8-0 45 0	ZV DV .40 .85	SLN 31/200 41.49 1.29		SLN 110 5.93 0.43	SLN 25/15 41.93 0.73	SLN 26/151 41.25 0.78	KVI 5/61 44.5 0.1	D         KVI           0         7/3           6         44.4           3         0.3	D K 0 27 4 4 4	VD 7/65 2.92 { 1.92	OZV Z-12 55.79 0.44	KVD MK-10 51.83 0.82		37.20 2.68
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub>	O e 8-0 45 0 3	ZV DV .40 .85 .87	SLN 31/200 41.49 1.29 14.31		SLN 110 5.93 0.43 9.03	SLN 25/15 41.93 0.73 15.94	SLN 26/151 41.25 0.78 16.20	KVI 5/61 44.5 0.1 19.5	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	$\begin{array}{c} D \\ 0 \\ 27 \\ 4 \\ 4 \\ 0 \end{array}$	VD 7/65 2.92 1.92 7.20	OZV Z-12 55.79 0.44 19.07	KVD MK-10 51.83 0.82 16.52		37.20 2.68 7.20
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub>	O e 8-0 45 0 3 3	ZV DV .40 .85 .87 .55	SLN 31/200 41.49 1.29 14.31 5.96		5.93 0.43 0.03 1.42	SLN 25/15 41.93 0.73 15.94 4.48	SLN 26/151 41.25 0.78 16.20 6.57	KVI 5/61 44.5 0.1 19.5 1.1	D         KVI           0         7/3           6         44.4           3         0.3           6         12.5           7         0.2	$\begin{array}{cccc}                                  $	VD 7/65 2.92 1.92 7.20 4.56	OZV Z-12 55.79 0.44 19.07 1.66	KVD MK-10 51.83 0.82 16.52 2.18		37.20 2.68 7.20 8.96
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO	O e 8-0 45 0 3 3 6	ZV DV .40 .85 .87 .55 .76	SLN 31/200 41.49 1.29 14.31 5.96 4.51		5.93 ).43 ).03 [.42 2.47	SLN           25/15           41.93           0.73           15.94           4.48           6.71	SLN 26/151 41.25 0.78 16.20 6.57 4.67	KVI 5/61 44.5 0.1 19.5 1.1 2.0	D         KVI           0         7/3           6         44.4           3         0.3           6         12.5           7         0.2           1         7.4	$\begin{array}{c ccc} \hline 0 & 27 \\ \hline 4 & 42 \\ \hline 4 & 5 \\ \hline 2 & 4 \\ 3 & 7 \\ \hline \end{array}$	VD 7/65 2.92 \$ 1.92 7.20 \$ 4.56 7.92	OZV Z-12 55.79 0.44 19.07 1.66 1.72	KVD MK-10 51.83 0.82 16.52 2.18 4.70		37.20 2.68 7.20 8.96 6.18
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO	O e 8-0 45 0 3 3 6 0	ZV 2V .40 .85 .87 .55 .76 .24	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16		SLN 110 5.93 0.03 1.42 2.47 0.06	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1	D         KVI           0         7/3           6         44.4           3         0.3           6         12.5           7         0.2           1         7.4           0         0.1	D K 0 27 4 4 4 2 4 3 7 8 0	VD 7/65 2.92 { 1.92 7.20 1 4.56 7.92 0.14	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05	KVD MK-10 51.83 0.82 16.52 2.18 4.70 0.12		37.20 2.68 7.20 8.96 6.18 0.25
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO	O e 8-0 45 0 3 3 6 0 8	ZV DV .40 .85 .87 .55 .76 .24 .97	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34	S 25/ 25/ 38 0 ( 19 19 19 19 19 19 19 19 19 19	5.93 0.03 1.42 2.47 0.06 4.77	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	D     K       0     27       4     4       0     4       2     4       3     4       6     1	VD 7/65 2.92 { 1.92 7.20 1 4.56 7.92 0.14 1.86	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61	KVD MK-10 51.83 0.82 16.52 2.18 4.70 0.12 3.38		37.20 2.68 7.20 8.96 6.18 0.25 12.62
Massif Sample SiO <sub>2</sub> TiO <sub>2</sub> Al <sub>2</sub> O <sub>3</sub> Fe <sub>2</sub> O <sub>3</sub> FeO MnO MgO CaO	O e 8-0 45 0 3 3 6 0 8 22	ZV DV .40 .85 .87 .55 .76 .24 .97 .67	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93	25/	5.93 0.43 0.03 1.42 2.47 0.06 4.77 3.44	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5	D         KVI           0         7/3           6         44.4           3         0.3           6         12.5           7         0.2           1         7.4           0         0.1           6         5.2           2         14.7	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 7.20 1.4.56 7.92 0.14 1.86 7.40	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11
Massif Sample SiO <sub>2</sub> 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	0 e 8-0 45 0 3 3 6 0 8 22 22 2	ZV DV .40 .85 .87 .55 .76 .24 .97 .67 .52	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88	S 25/ 25/ 38 0 (19 19 19 19 19 19 19 11 11 11	5.93 0.43 0.03 1.42 2.47 0.06 1.77 3.44 1.44	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 1.4 56 7.92 0.14 1.86 7.40 2.67	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97
Massif Sample SiO <sub>2</sub> 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	$\begin{array}{c c} & & \\ & & \\ \hline & & \\ &$	ZV DV .40 .85 .87 .55 .76 .24 .97 .67 .52 .34	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82	S         S           25/         33           0         12           1         12           2         13           3         13           3         13           3         13	5.93 ).43 ).03 1.42 2.47 ).06 4.77 3.44 1.44 3.04	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9 4.0	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 1.4.56 7.92 0.14 1.86 7.40 2.67 2.67 2.67 2.67 2.67 2.64	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ Ma_2O \\ K_2O \\ P_2O_5 \end{tabular}$	$\begin{array}{c c} & & \\ & & \\ \hline & & \\ &$	ZV DV .40 .85 .87 .55 .76 .24 .97 .67 .52 .34 .41	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89	$\begin{array}{c c} & & & \\ & & & & \\ & & & \\ & & & & \\ & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\$	5.93 0.43 0.03 1.42 2.47 0.06 4.77 3.44 1.44 3.04 5.36	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9 4.0 0.7	$\begin{array}{c cccc} \hline & & & & \\ \hline & & & & \\ \hline & & & & \\ \hline & & & &$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 1.56 7.92 0.14 1.86 7.40 2.67 2.67 2.67 2.64 0.20	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ Ma_2O \\ Ma_2O \\ K_2O \\ P_2O_5 \\ CO_2 \end{tabular}$	$\begin{array}{c c} & & \\ & &$	ZV 2V 2V 2V 2V 2V 2V 2V 2V 2V 2	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89 0.43	$\begin{array}{c c} & & & \\ & & & \\ \hline & & \\ \hline & & & \\ \hline \\ \hline$	5.93 ().43 ().03 ().42 ().03 ().42 ().03 ().42 ().42 ().44 ().44 ().44 ().44 ().44 ().44 ().66 ().19	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76           n.a.	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67 0.59	KVI 5/610 44.50 0.11 19.55 1.11 2.00 0.10 5.60 9.55 9.99 4.00 0.77 0.8	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 1.56 7.92 0.14 1.86 7.40 2.67 2.67 2.64 0.20 0.80	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21 1.89	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54 0.54		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20 1.76
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ Ma_2O \\ MgO \\ CaO \\ Na_2O \\ K_2O \\ P_2O_5 \\ CO_2 \\ S_{tot}. \end{tabular}$	$\begin{array}{c c} & & \\ & & \\ \hline \\ & & \\$	ZV DV .40 .85 .87 .55 .76 .24 .97 .67 .52 .34 .41 .26 .01	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89 0.43 0.10	25/25/	SLN 110 5.93 0.43 0.03 1.42 2.47 0.06 1.77 3.44 1.44 3.04 5.36 1.19 0.30	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76           n.a.           0.04	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67 0.59 0.05	KVI 5/610 44.50 0.11 19.55 1.11 2.00 0.10 5.60 9.55 9.99 4.00 0.77 0.8 0.0	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 1.56 7.92 0.14 1.86 7.40 2.67 2.67 2.64 0.20 0.80 n.a.	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21 1.89 n.a.	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54 0.54 0.44 0.10		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20 1.76 -
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ MgO \\ CaO \\ Na_2O \\ K_2O \\ P_2O_5 \\ CO_2 \\ S_{tot.} \\ F \end{tabular}$	$\begin{array}{c c} & & \\ & & \\ \hline \\ & & \\$	ZV 2V 2V 2V 2V 2V 2V 2V 2V 2V 2	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89 0.43 0.10 0.06	$\begin{array}{c c} & & & \\ \hline \\ \hline$	5.93 0.43 0.03 1.42 2.47 0.06 4.77 3.44 1.44 3.04 5.36 1.19 0.30 0.27	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76           n.a.           0.04           0.04	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67 0.59 0.05 0.03	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9 4.0 0.7 0.8 0.0 0.0	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 1.92 7.20 1.56 7.92 0.14 1.86 7.40 2.67 2.67 2.064 0.20 0.80 n.a. 0.06	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21 1.89 n.a. 0.03	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54 0.54 0.44 0.10 0.10		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20 1.76 
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ Ma_2O \\ Ma_2O \\ K_2O \\ P_2O_5 \\ CO_2 \\ S_{tot.} \\ F \\ LOI \end{tabular}$	$\begin{array}{c c} & & \\ & & \\ \hline \\ & & \\ & \\ & \\ & \\ & \\ &$	ZV DV .40 .85 .87 .55 .76 .24 .97 .67 .52 .34 .41 .26 .01 .20 .62	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89 0.43 0.10 0.06 0.66	$\begin{array}{c c} & & & \\ \hline \\ \hline$	SLN 110 5.93 0.43 0.03 1.42 2.47 0.06 4.77 3.44 1.44 3.04 5.36 1.19 0.30 0.27 1.34	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76           n.a.           0.04           0.85	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67 0.59 0.05 0.03 0.41	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9 4.0 0.7 0.8 0.0 0.0 1.1	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 2.92 1.92 7.20 4.56 7.92 0.14 1.86 7.40 2.67 2.67 2.07 2.67 2.07 2.07 2.07 2.00 1.92 0.14 1.86 7.40 0.267 2.00 1.92 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.20 0.14 1.86 0.04 0.20 0.14 1.80 0.20 0.14 1.80 0.04 0.20 0.14 1.80 0.20 0.14 1.80 0.04 0.20 0.14 1.80 0.04 0.20 0.14 0.20 0.14 1.80 0.04 0.20 0.14 1.80 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0.14 0.20 0	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21 1.89 n.a. 0.03 1.65	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54 0.44 0.10 0.10 1.86		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20 1.76 
$\begin{tabular}{ c c c c c } \hline Massif \\ \hline Sample \\ \hline SiO_2 \\ TiO_2 \\ Al_2O_3 \\ Fe_2O_3 \\ Fe_2O_3 \\ FeO \\ MnO \\ MgO \\ CaO \\ Ma_2O \\ Ma_2O \\ K_2O \\ P_2O_5 \\ CO_2 \\ S_{tot.} \\ F \\ LOI \\ Total \end{tabular}$	O e 8-0 45 0 3 3 3 6 6 0 8 22 2 2 0 3 3 0 0 0 0 0 0 99	ZV 2V 2V 2V 2V 2V 2V 2V 2V 2V 2	SLN 31/200 41.49 1.29 14.31 5.96 4.51 0.16 5.34 12.93 7.88 2.82 1.89 0.43 0.10 0.06 0.66 99.83	$\begin{array}{c c} & & & \\ & & & \\ \hline & & \\ \hline & & & \\ \hline \\ \hline$	SLN 110 5.93 0.43 0.03 1.42 2.47 0.06 1.77 3.44 1.44 3.04 5.36 1.19 0.30 0.27 1.34 0.49	SLN           25/15           41.93           0.73           15.94           4.48           6.71           0.17           5.35           11.16           8.53           2.84           0.76           n.a.           0.04           0.85           99.53	SLN 26/151 41.25 0.78 16.20 6.57 4.67 0.13 5.62 11.30 8.77 2.99 0.67 0.59 0.05 0.03 0.41 100.03	KVI 5/61 44.5 0.1 19.5 1.1 2.0 0.1 5.6 9.5 9.9 4.0 0.7 0.8 0.0 0.0 1.1 99.5	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	VD 7/65 1.92 7.20 1.92 7.20 1.4.56 7.92 0.14 1.86 7.40 2.67 1.20 0.64 0.20 0.80 n.a. 0.06 1.29 9.58 9	OZV Z-12 55.79 0.44 19.07 1.66 1.72 0.05 0.61 2.39 11.35 2.43 0.21 1.89 n.a. 0.03 1.65 99.29	KVD KVD 51.83 0.82 16.52 2.18 4.70 0.12 3.38 4.44 8.58 3.64 0.54 0.44 0.10 0.10 1.86 99.25		37.20 2.68 7.20 8.96 6.18 0.25 12.62 17.11 2.97 1.87 1.20 1.76 - - 00.00

Note: Hereinafter in Tables: n.a. – not analyzed, n.d. – not detected,  $mg\# = Mg/(Mg+Fe^{2+})$ . Rock symbols: DUN – dunite, PRD – peridotite, PRX – pyroxenite, MEL – melilitolite, MELT – turjaite, MLG – melteigite, IJL – ijolite, NFS – nepheline syenite. AVER<sup>\*</sup> – weighted mean composition of the alkaline ultramafic series of the Kola province, after *Arzamastsev et al.* [2001].

[Kravchenko et al., 1979] has not been confirmed by any single measurement. Comparison with other provinces in terms of REE contents and patterns shows that apatite compositions under study resemble those from carbonatites of the Alno, Sokli, and Fen massifs [Hornig-Kjarsgaard, 1998].

*Titanite.* We have analyzed late titanites, which replace (Table 7). All

perovskite in Afrikanda pyroxenites (sample 25-AFR), and groundmass titanites from Ozernaya Varaka and Khibiny ijolites. Comparison shows that Khibiny varieties are appreciably enriched in Sr, whereas Ozernaya Varaka and Afrikanda titanites have elevated concentrations of Zr, Hf, U, and Th (Table 7). All the varieties have narrow ranges of REE con-

Rock	PRD	PRD	PRX	PRX	PRX	MEL	MEL	MLG	MLG	MLG	MLG	IJL	IJL	IJL	IJL
Sample	A-1044	1010/	A-1036	A-1087	A-1038	1010/	1010/	1119/	455/	466/	1010/	1072/	455/	1636/	1152/
		1052				1059	1165	279	345	552	1186	530	402	721	83
$SiO_2$	43.62	38.26	41.60	37.53	41.37	28.01	29.98	41.24	37.56	44.00	39.52	43.48	44.81	43.67	46.07
TiO <sub>2</sub>	3.41	3.51	4.91	2.90	4.18	6.01	2.84	6.42	7.73	4.60	7.25	2.67	1.97	1.61	2.28
$Al_2O_3$	5.21	4.29	5.92	11.09	7.37	6.12	3.58	4.87	3.38	3.64	9.17	18.85	17.44	21.03	17.84
$Fe_2O_3$	5.41	10.31	4.33	5.90	8.90	15.57	10.51	9.23	11.60	11.72	8.00	4.88	5.31	4.53	5.17
FeO	7.06	5.25	11.25	4.04	5.22	5.83	5.18	8.22	11.81	7.70	7.00	3.17	3.19	3.95	2.90
MnO	0.19	0.28	0.23	0.17	0.23	0.26	0.29	0.48	0.61	0.33	0.28	0.16	0.21	0.22	0.29
MgO	17.85	19.93	11.67	6.49	13.38	9.20	16.58	5.74	7.17	8.32	7.01	3.35	3.94	1.54	1.79
CaO	9.36	12.60	11.20	18.40	9.09	23.43	28.46	13.61	14.68	13.41	11.74	7.70	7.91	4.63	3.94
Na <sub>2</sub> O	2.26	0.65	3.47	5.69	3.18	1.35	0.67	4.93	3.30	4.02	4.56	10.58	10.74	13.21	11.58
$K_2O$	2.91	2.06	2.11	1.78	3.60	0.86	0.04	1.73	0.81	1.46	2.36	4.45	3.78	3.42	5.20
$P_2O_5$	0.36	0.42	0.38	2.97	0.36	0.77	0.37	1.88	0.99	0.27	1.25	0.65	0.22	1.00	0.37
$CO_2$	0.14	0.38	0.23	0.96	0.32	0.49	0.35	n.a.	0.05	0.05	0.32	0.31	0.07	0.15	0.08
S <sub>tot</sub> .	0.03	0.16	0.12	0.40	0.04	0.04	0.02	0.30	n.a.	n.a.	0.19	n.a.	n.a.	0.04	n.a.
F	0.45	1.06	0.60	0.14	0.91	0.58	0.16	0.43	0.20	0.36	0.32	0.09	0.07	0.21	0.26
$H_2O$	1.01	0.68	1.52	1.27	1.32	1.53	0.46	0.72	0.33	1.08	0.48	0.20	0.48	0.70	1.23
Total	99.27	99.84	99.54	99.73	99.47	100.05	99.49	99.80	100.22	100.96	99.45	100.54	100.14	99.91	99.00
mg#	0.82	0.87	0.65	0.74	0.82	0.74	0.85	0.38	0.52	0.66	0.64	0.65	0.69	0.25	0.29

**Table 3.** Whole-rock major element analyses (wt %) for the Khibiny alkaline ultramafic series

tents and relatively low  $(La/Yb)_N$  ratios (40–53), which results in gently sloping, straight chondrite-normalized patterns (Figure 7).

5.2.2. Rock-forming minerals. Olivine. To analyze trace element contents, we selected the purest olivine grains of Fo<sub>82-91</sub> composition from Kovdor and Lesnava Varaka dunites. This, however, did not ensure the absence of magnetite microinclusions, which are distributed evenly within olivine crystals. Scanning electron microscopy detects extremely thin perovskite rims around primary chromite inclusions in olivine. On the one hand, positive correlation of Nb and Ta with REE in olivine (Table 8) implies the presence of perovskite microinclusions. On the other hand, all the olivine specimens analyzed show negative Eu anomaly, also observable in magnetite that coexists with olivine (Figure 8). The highest Eu oxidation degree  $(Eu/Eu^* < 0.08)$ was established in Lesnaya Varaka dunites, which contain significant amounts of magnetite. Based on this evidence, we assume that a considerable fraction of REE contained in olivine grains is concentrated not in this mineral proper, but in perovskite and magnetite microinclusions.

Clinopyroxene. In early cumulates of the alkaline ultramafic series, clinopyroxene is represented by diopside  $Di_{80}Hd_{15}Ac_5$ , with foidolites and nepheline syenites containing zoned segregations that range in composition from augite  $Di_{55}Hd_{40}Ac_5$  to aegirine-augite  $Di_{50}Hd_{30}Ac_{20}$  [Arzamastsev and Arzamastseva, 1993; Kukharenko et al., 1965]. A study of zoned pyroxene grains using the laser ablation technique [Arzamastsev et al., 2001b] shows that zone-to-zone variations of major elements within crystals are not accompanied by any marked variations in the contents of trace elements, in particular, REE. This is further supported by analyses of pyroxene separates from pyroxenites, melilitolites, and ijolites from carbonatite massifs of the province, which reveal a narrow range of REE variations in all the groups of rocks (Table 8). The higher REE contents are observed in rock varieties from Khibiny ijolites, although all the rocks have the same REE distribution pattern. Plots shown in Figure 8 demonstrate all the clinopyroxenes to be enriched in Yb and Lu relative to Dy, Ho, and Er.

Melilite. Microprobe measurements on melilites from rocks of the Kovdor massif yield the following composition (mol %): Mg-akermanite, 49–70; Fe-akermanite, 5–11; Namelilite, 36–40. Compared with previously published data on melilite compositions in the rocks of the province [Bell et al., 1996], the analyzed melilites have higher Mg/Fe ratios, a feature typical of Ne-free alkaline ultramafites [Mitchell, 1996a; Rass, 1986]. Overall, the melilite is high in Sr due to the isomorphous replacement  $\mathrm{Sr}^{2+} \rightarrow \mathrm{Ca}^{2+}$ . Unlike Khibiny melilites, Kovdor ones display no appreciable enrichment in Sr (Table 8). Compared with the scanty reported REE measurements on melilites from various provinces [Mitchell, 2001; Onuma et al., 1981], Kovdor specimens have somewhat lowered total REE contents and straight patterns with relatively high (La/Yb)<sub>N</sub> ratios of 178–203 (Figure 8).

Nepheline. Nepheline compositions in rocks of the Kovdor-type alkaline ultramafic intrusions range from Ne<sub>77.8-82.5</sub>Ks<sub>9.6-18.9</sub>Qz<sub>1.4-3.2</sub> to Ne<sub>78.6-81.6</sub>Ks<sub>15.2-20.1</sub> Qz<sub>1.4-3.2</sub> in foidolites and Ne-syenites, respectively. Nephelines from Khibiny foidolites are higher in the kalsilite end-member and silica (Ne<sub>68.4-72.8</sub>Ks<sub>21.5-26.1</sub>Qz<sub>4.8-7.4</sub>). Because of high Fe<sub>2</sub>O<sub>3</sub> contents in the matrix of nepheline, all the nepheline grains are replete with aegirine microlites representing exsolution products. REE contents in all the nepheline varieties range from low to very low (Table 8). Chondrite-normalized REE patterns are straight (Figure 8), nephelines from Khibiny ijolites having negative Eu anomaly

Rock	DUN	DUN	DUN	PRD	PRX	PRX	PRX	PRX	PRX	PRX	MELT	MELT	MLG	MLG	MLG
Massif	KVD	LSV	SLM	AFR	AFR	AFR	VUO	VUO	VUO	SLM	KVD	KVD	KVD	KVD	KVD
Sample	4/400	Z-6	SL-23	AFR-1	AFR-5	AFR-6	93/192	282/224	282/251	SL-24	227/43	252/100	7/394	7/348	5/740
Li	n.d.	2.58	3.32	2.11	n.d.	4.80	n.d.	0.81	1.86	6.22	n.d.	n.d.	n.d.	n.d.	n.d.
Rb	4.53	0.94	18.7	1.54	5.05	25.0	10.4	18.9	11.69	11.7	47.3	35.8	69.6	23.4	63.0
Cs	n.d.	0.08	0.29	0.05	n.d.	0.11	n.d.	0.24	0.19	0.14	0.36	0.20	0.61	0.12	0.75
Be	0.18	0.26	0.86	0.47	2.37	6.20	1.73	1.29	1.12	7.03	8.42	9.19	1.81	2.62	1.90
Sr	74.5	7.67	160	766	1525	1313	399	677	369	1103	1998	3050	358	563	298
Ba	21.1	12.3	152	30.7	20.6	25.7	151	297	222	801	1011	951	401	103	937
Sc	7.82	7.09	8.80	31.3	14.9	14.4	79.0	89.5	43.0	46.2	12.6	2.34	28.5	97.5	13.5
V	20.8	20.8	58.3	190	177	210	142	202	375	355	73.1	41.6	171	191	190
Cr	1462	2124	2725	27.3	23.2	10.2	30.3	116	171	38.6	3.96	5.07	1083	51.1	30.4
Co	112	168	93.9	96.4	35.8	69.0	35.8	46.1	85.1	52.4	36.7	35.1	49.6	12.9	69.1
Ni	1480	1432	1961	54.5	40.6	29.7	59.8	75.3	148	54.5	145	85.4	321	21.5	231
Cu	18.1	7.58	25.8	49.9	314	216	34.5	58.1	33.8	415	17.6	26.1	162	12.3	28.2
Zn	70.0	79.9	85.9	201	34.2	108	48.3	144	208	148	183	219	79.8	44.1	194
Ga	3.84	1.91	2.68	27.5	12.9	28.1	16.9	26.5	31.7	18.3	15.9	15.9	14.8	5.27	16.5
Y	1.49	0.69	2.38	49.3	66.9	25.9	27.9	78.5	36.3	29.4	3.18	3.76	6.20	10.3	3.51
Nb	8.64	3.00	10.3	1412	31.6	379	208	818	561	238.3	12.7	10.8	18.8	18.5	32.4
Ta	0.41	0.38	0.44	132	1.89	41.7	20.4	42.2	48.5	13.7	0.86	0.73	0.46	0.93	2.80
Zr	21.0	6.31	80.9	111	419	472	456	450	482	425	98.5	25.8	113	190	210
Hf	0.55	n.d.	1.93	4.75	12.9	5.72	18.0	15.9	17.3	11.9	2.88	0.68	2.99	4.78	2.59
Sn	2.18	0.78	n.d.	7.38	4.96	5.21	6.37	7.79	11.0	7.37	2.71	1.83	3.43	3.93	4.99
Pb	0.98	0.71	0.48	5.43	13.0	5.93	1.65	3.16	1.48	4.15	2.50	3.53	5.05	2.14	1.73
U	0.29	n.d.	0.43	16.2	1.53	6.19	3.28	14.5	6.54	5.38	0.22	0.24	0.45	0.50	0.83
Th	0.95	0.29	0.48	201	7.36	72.7	15.9	87.3	43.0	12.2	0.81	0.88	1.07	3.80	2.30
La	5.16	0.95	5.39	1724	354	475	224	1046	513	172	28.5	42.6	21.2	42.5	8.25
Ce	11.1	1.66	10.1	4827	551	1037	502	2800	1343	368	52.9	76.2	50.0	99.6	18.5
Pr	1.24	0.17	1.13	359	53.0	111	57.5	256	144	43.2	5.71	8.08	6.09	12.2	2.32
Nd	3.37	0.55	4.08	1131	189	386	222	888	516	142	19.9	27.4	22.1	47.7	8.14
Sm	0.60	0.11	0.70	124	31.1	44.2	35.5	125	70.0	20.2	2.69	3.74	3.57	7.30	1.58
Eu	0.23	0.02	0.21	29.4	9.23	10.1	9.26	33.1	17.7	5.36	0.52	0.84	0.95	1.99	0.36
Gd	0.32	0.08	0.65	78.9	24.8	23.2	24.1	86.9	47.3	13.8	1.62	2.06	2.28	5.15	1.02
Tb	0.05	0.02	0.09	8.57	3.18	2.31	2.70	10.1	5.35	1.74	0.16	0.22	0.28	0.59	0.14
Dy	0.32	0.09	0.47	24.3	16.0	8.19	10.9	32.1	17.6	7.64	0.71	0.90	1.42	2.59	0.77
Ho	0.06	0.02	0.10	2.96	2.74	1.10	1.52	4.52	2.26	1.23	0.13	0.14	0.26	0.41	0.14
Er	0.19	0.05	0.26	4.72	5.92	2.14	2.65	7.55	3.76	2.70	0.28	0.32	0.63	0.90	0.40
Tm	0.03	0.01	0.04	0.43	0.67	0.25	0.31	0.75	0.36	0.36	0.04	0.04	0.10	0.13	0.07
Yb	0.21	0.06	0.24	2.23	3.50	1.48	1.67	3.65	1.80	2.03	0.25	0.23	0.67	0.93	0.42
Lu	0.03	0.01	0.04	0.23	0.49	0.21	0.23	0.41	0.21	0.27	0.04	0.03	0.12	0.14	0.08
$ (La/Yb)_N $	17.4	11.7	15.7	521	71.1	226	94.2	193	192	59.6	81.9	128	22.3	32.0	13.9

Table 4. Whole-rock trace element analyses (ppm) for carbonatite-bearing alkaline ultramafic massifs of the Kola province

 $(Eu/Eu^* = 0.13)$ , evidently due to the presence of magnetite microinclusions, rather than aegirine ones only. On the other hand, in nephelines from Kovdor ijolites, which are least abundant in aegirine microlites, Eu anomaly is expressed poorly  $(Eu/Eu^* = 0.69)$ .

**5.2.3. REE** distribution in coexisting mineral phases. Data on REE distribution in principal REE- bearing phases—perovskite, apatite, and titanite (Tables 6, 7)— enable us to calculate partition coefficients for these phases.

Comparison of coefficients obtained for the coexisting pairs perovskite/apatite ( $D_{Prv}/A_{P}$ ), perovskite/titanite ( $D_{Prv}/Tit$ ), and apatite/titanite ( $D_{Ap/Tit}$ ) (Table 9) shows that in early pyroxene–perovskite cumulates, REE partition preferentially into the perovskite. Overall, during magmatic crystallization, REE enter the above minerals in the following order: perovskite > apatite > titanite. According to our calculations, the early-generation perovskite (sample AFR-5) extracts medium- and heavy REE most strongly ( $D_{Prv}/A_{P}$ for Tb–Lu > 3), whereas in perovskite IIsapatite pairs (sam-

Table 4. Continued

Rock	MLG	IJL	IJL	IJL	IJL	IJL	IJL	IJL	SFN	SFN	$AVER^*$
Massif	OZV	SLN	SLN	SLN	SLN	KVD	KVD	KVD	OZV	KVD	
Sample	8-OV	31/200	25/110	25/15	26/151	5/610	7/30	27/65	Z-12	MK-10	
Li	4.20	3.71	2.92	n.d.	n.d.	n.d.	n.d.	5.21	4.36	43.4	13.5
Rb	7.91	46.7	42.7	50.1	52.9	54.6	63.7	15.5	20.0	131	49.3
Cs	0.12	0.30	0.24	0.04	n.d.	0.04	0.01	n.d.	2.91	0.94	0.64
Be	2.56	2.90	2.40	1.78	1.72	6.52	10.0	1.29	4.53	4.27	_
Sr	1106	561	879	359	388	281	730	429	967	1039	2094
Ba	31.6	88.6	87.9	76.2	67.1	129	163	210	956	1101	1116
Sc	11.9	4.75	2.89	8.01	13.5	22.3	11.6	49.7	1.11	10.7	24.8
V	200	175	46.7	101	94.9	85.4	175	294	66.6	124	104
Cr	59.1	46.0	15.3	13.3	19.6	5.16	76.0	409	12.4	55.4	293
Co	32.5	31.6	11.8	31.6	33.1	7.55	23.0	53.7	3.38	18.9	42.5
Ni	24.4	16.7	7.73	19.4	18.2	7.55	30.8	130	1.99	31.9	243
Cu	12.0	135	65.0	62.6	118	9.05	23.4	205	4.60	58.4	-
Zn	75.1	70.7	10.4	59.4	58.5	22.7	96.0	61.9	18.0	102	-
Ga	10.6	21.7	18.0	18.3	19.5	20.1	30.7	13.1	17.4	20.8	_
Y	20.5	19.3	26.7	6.25	6.33	4.66	7.21	10.6	1.96	12.5	33.6
Nb	29.0	48.0	18.6	16.3	16.3	20.4	65.3	16.1	135	52.8	95.8
Ta	2.05	5.53	1.52	0.54	0.81	2.16	3.99	0.73	10.2	2.87	5.50
Zr	164	182	74.4	58.2	55.4	264	283	82.8	146	155	347
Hf	3.55	4.36	2.07	1.71	1.80	5.41	4.79	3.24	2.53	3.73	7.65
Sn	2.42	3.70	3.91	2.43	2.69	3.12	3.19	3.54	4.87	3.30	_
Pb	1.59	1.97	10.6	2.69	3.39	1.27	2.53	4.13	1.53	13.3	-
U	0.83	0.40	1.03	0.49	0.44	0.50	1.54	0.38	0.72	1.68	2.15
Th	12.2	4.36	10.3	1.86	2.33	7.02	17.2	1.31	0.75	8.56	8.51
La	169	38.6	113	21.6	22.6	32.0	55.4	18.5	12.8	52.1	131
Ce	315	83.1	216	40.0	44.1	70.2	118	36.2	30.0	88.7	249
Pr	34.1	9.86	23.7	4.42	4.99	7.93	13.1	4.19	3.66	8.40	27.4
Nd	116	38.2	86.3	15.5	18.3	28.7	46.0	16.1	12.8	27.3	96.9
Sm	15.1	6.65	12.7	2.55	3.12	4.22	5.94	2.92	1.77	3.99	14.4
Eu	3.60	2.01	3.63	0.80	0.91	1.12	1.45	0.97	0.49	0.78	3.60
Gd	11.1	6.02	10.9	2.06	2.33	2.69	3.47	2.52	1.34	2.84	10.2
Tb	1.28	0.87	1.39	0.25	0.28	0.30	0.37	0.37	0.17	0.39	1.26
Dy	5.26	4.84	6.01	1.31	1.38	1.32	1.48	2.07	0.68	2.16	6.02
Ho	0.82	0.84	1.11	0.25	0.25	0.20	0.27	0.40	0.10	0.44	1.05
Er	1.87	2.03	2.50	0.58	0.57	0.42	0.71	0.99	0.20	1.26	2.42
Tm	0.25	0.30	0.32	0.09	0.08	0.06	0.10	0.15	0.03	0.19	0.32
Yb	1.50	1.77	1.71	0.60	0.54	0.50	0.72	0.93	0.17	1.22	1.83
Lu	0.25	0.30	0.23	0.11	0.10	0.08	0.12	0.14	0.03	0.18	0.26
$(La/Yb)_N$	76.2	14.7	44.6	25.4	29.5	45.0	54.3	14.0	51.2	30.0	48.4

ple 8-OV) the bulk of the heavy REE partition into the apatite. The results obtained are consistent with microprobe measurements on minerals from Ugandan clinopyroxenites and kamafugite lavas [*Lloyd et al.*, 1996] and from the plutonic Oldoinyo Lengai alkaline ultramafic rocks [*Dawson et al.*, 1994, 1995], which reveal a relative constancy of REE distribution in the perovskite–apatite pair (D<sub>Prv/Ap</sub>: La 9, Ce 16, Nd 9.5). D<sub>Prv/Ap</sub> values similar to those obtained by us for the late perovskite II–apatite pairs were established for Kaiserstuhl calcite carbonatites [*Hornig-Kjarsgaard*, 1998], D<sub>Prv/Ap</sub> in these rocks decreasing from 6.8–4.9 for the LREE to 3.0-1.5 for the MREE through to 0.5 for Yb.

5.2.4. Estimating REE distribution between minerals and rocks. In view of the fact that alkaline ultramafic rocks under study are differentiates, and their chemistries do not necessarily correspond to compositions of those melts that precipitated the mineral phases contained in these rocks, estimates of mineral/melt partition coefficients (D<sub>mineral</sub>) may be very tentative. Hence, when calculating D REE for early phases from pyroxene and olivine

IJL

1152/83 44.297.6 0.9515.3328026095.5020883.4 14.864.383.220038.256.7387 11.2213132.3

Rock	PRD	PRD	PRX	PRX	PRX	MEL	MEL	MLG	MLG	MLG	MLG	IJL	IJL	IJL
Sample	A-1044	1010/1052	A-1036	A-1087	A-1038	$1010/\\1059$	$1010/\\1165$	$\frac{1119}{279}$	$\frac{455}{345}$	$rac{466}{552}$	1010/ 1186	$\frac{1072}{530}$	455/ 402	1636/ 721
Li	36.6	8.0	46.1	2.6	84.3	6.00	7.50	32.4	12.2	28.4	22.3	28.5	7.90	14.5
Rb	115.4	138.4	112.7	23.9	207.8	58.5	1.10	51.8	15.9	37.7	86.0	23.4	34.4	71.3
Cs	1.30	2.60	1.52	0.32	2.20	0.79	0.03	0.55	0.32	0.22	0.60	3.62	0.19	0.78
Be	0.99	2.54	7.39	4.68	5.46	5.61	3.17	7.36	3.02	3.34	4.25	6.53	5.73	5.70
Sr	893	1126	995	748	1485	5933	539	6168	2544	3237	2194	31.3	1352	4974
Ba	558	1364	432	187	521	1063	49.4	1166	168	1114	852	357	335	49.2
$\mathbf{Sc}$	21.4	23.9	28.2	19.0	21.1	42.0	24.0	23.3	26.5	22.3	20.0	2.50	19.5	2.67
V	213	506	251	242	265	298	83.1	547	673	739	428	1.80	276	322
$\operatorname{Cr}$	1131	1277	800	22.7	724	158	1061	92.7	112	39.1	135	72.8	51.4	38.9
Co	80.7	98.0	66.1	27.7	62.5	83.8	102	49.8	65.1	75.6	45.9	2.6	16.0	10.9
Ni	915	1018	428	37.8	568	278	994	77.2	104	61.9	134	37.4	29.1	5.61
Cu	65.2	963	39.4	237	26.1	194	34	285	534	628	455	10.3	18.0	21.0
Zn	157	190	305	131	226	277	159	247	429	143	325	80.8	105	102
Ga	12.2	12.4	28.2	21.1	24.2	33.4	9.2	20.0	14.3	12.8	24.6	32.3	32.9	37.0
Y	19.7	27.7	30.4	45.9	33.6	30.5	18.3	76.8	63.0	36.0	86.6	41.8	24.1	21.4
Nb	39.0	73.4	104	154	94.8	312	157	304	261	169	371	215	110	101
Ta	2.55	5.41	7.64	6.83	5.75	23.3	15.1	17.8	31.2	16.4	28.0	15.8	7.98	5.59
Zr	220	345	481	429	453	480	173	1168	812	814	744	929	565	564
Hf	5.94	8.74	10.3	12.2	11.1	12.5	4.59	23.4	19.8	24.8	18.0	25.3	12.9	11.5

Table 5. Whole-rock major element analyses (wt %) for the Khibiny alkaline ultramafic massif

cumulates, REE contents were correlated not with their host rocks, but with the mean composition for alkaline ultramafic rocks (Table 4), which best approximates composition of the melt parental to the series. Coefficients thus obtained are listed in Table 10 and are compared to published experimental values. In the plot showing  $D_{Prv/melt}$  for the entire REE spectrum (Figure 9), partition coefficients for Kola perovskites, while plotting somewhat lower, stay nonetheless with the same trend as does  $D_{Prv/melt}$  in melilite–olivine basalts, as determined by *Onuma et al.* [1981].

Sn

Pb

Th

La

Ce

 $\mathbf{Pr}$ 

Nd

Sm

Eu

Gd

 $^{\mathrm{Tb}}$ 

Dy

Ho

 $\mathbf{Er}$ 

Τm

Yb

Lu

 $(La/Yb)_N$ 

U

0.34

n.a.

0.41

2.67

62.0

138

16.5

68.4

11.6

3.29

8.47

1.03

4.99

0.81

1.85

0.21

1.21

0.17

36.0

1.96

n.a.

1.56

5.04

85.6

176

20.3

79.9

13.7

3.92

10.8

1.39

6.42

1.04

2.45

0.28

1.54

0.21

39.1

2.73

n.a.

1.76

9.09

102

223

26.4

105

18.0

5.08

13.2

1.65

7.57

1.22

2.63

0.31

1.82

0.27

39.3

3.10

n.a.

3.05

6.10

143

255

28.1

112

22.0

7.11

20.6

2.82

12.23

1.88

4.03

0.44

2.40

0.32

42.0

2.01

n.a.

1.86

7.22

102

216

24.8

99.0

16.7

4.70

12.0

1.64

7.71

1.34

3.04

0.37

2.19

0.30

32.8

5.92

n.a.

4.45

27.9

251

465

49.2

173

23.3

6.18

13.8

1.66

7.85

1.18

2.67

0.29

1.66

0.22

106.8

1.40

n.a.

2.97

19.2

128

270

28.9

103

13.3

3.47

7.50

0.90

4.20

0.69

1.64

0.20

1.20

0.18

75.2

7.53

6.42

4.44

16.9

330

652

66.8

235

35.0

10.2

29.4

3.98

17.9

3.09

6.94

0.93

5.17

0.74

48.1

5.00

n.a.

0.85

3.99

222

441

50.8

190

30.4

8.97

22.8

3.08

14.6

2.46

5.61

0.66

4.06

0.59

38.5

4.19

0.78

0.20

0.85

108

250

30.7

116

19.9

5.77

13.8

1.93

9.08

1.61

3.62

0.49

3.23

0.55

23.6

5.40

n.a.

1.57

23.4

380

747

88.9

323

49.5

14.3

33.1

4.37

21.0

3.42

7.83

0.86

4.80

0.60

55.7

8.12 2.17

n.a.

1.08

3.19

150

10.4

1.05

5.26

0.93

2.37

0.32

2.20

0.34

 $17.8 \ 22.2$ 

8.21

3.68

23.2

119 69.4

238

97.6 62.2

16.8

 $1.68 \ 3.04$ 

1.73

9.44

1.97

5.31

0.81

4.71

0.70

27.3 16.6

11.2 7.63

11.5

5.79

1.45

5.30

204

332

30.3

96.1

12.0

3.24

9.36

1.20

4.88

0.86

1.97

0.29

1.95

0.34

70.4

6.31

11.9

8.51

19.6

116

200

19.7

67.3

11.1

3.40

10.5

1.68

9.23

2.01

5.63

0.86

5.05

0.71

15.4

In calculations of D REE for minerals from the more

evolved members of the alkaline ultramafic series (ijolites), we assumed their constituent phases (clinopyroxene, melilite, apatite, and titanite) to have been in equilibrium with the host rocks. The obtained  $D_{Cpx/rock}$  and  $D_{Ap/rock}$  values, listed in Table 11, broadly compare to those for minerals in alkaline volcanites from other regions [*Caroff et al.*, 1993; *Foley et al.*, 1996; *Irving and Price*, 1981; *Larsen*, 1979; *Onuma et al.*, 1981]. On the other hand, partition coefficients for titanites from Khibiny and Ozernaya Varaka ijolites are lower than those for phonolite-hosted titanites [*Worner et al.*, 1983].



Figure 6. Chondrite-normalized REE patterns for alkaline ultramafic rocks of the Kovdor and Khibiny types. In all the plots, heavy line shows REE pattern for the weighted mean composition of the alkaline ultramafic series. Normalizing values, after [Anders and Grevesse, 1989].

+

# 6. Discussion

Petrologic evidence, supported by experimental data [Edgar, 1987; Le Bas, 1987; Onuma and Yamamoto, 1976; Pan and Longhi, 1989, 1990; Veksler et al., 1998a; Wilkinson and Stolz, 1983], indicate that the main process responsible for the genesis of alkaline ultramafic suites of various provinces worldwide was fractional crystallization of primary olivine melanephelinitic magma. Coherence of the rock formation sequence from early olivine- and clinopyroxene cumulates to melilitolite, foidolite, and Ne-syenite, is corroborated by geologic and petrographic observations. Estimates of magma compositions for the Kola alkaline province [Arzamastsev et al., 2001a] allow the inference that alka-

line ultramafics of the Kovdor and Khibiny types originated from the same primary magma. This is further supported by isotope studies, indicative of a single mantle source for Khibiny and Kovdor alkaline rocks [*Kramm and Kogarko*, 1994]. Variation trends in major-element plots for both suites reflect sequential precipitation of olivine, clinopyroxene, and melilite. Foidolites are produced through a reaction that involves resorption of melilite and formation of diopside and nepheline:



Figure 7. Chondrite-normalized REE patterns for perovskite, apatite, and titanite from alkaline ultramafic rocks. Normalizing values, from [Anders and Grevesse, 1989].

Published data concerning distribution of trace elements in the above principal mineral phases of the alkaline ultramafic series are indicative of REE, Sr, Y, Zr, Hf, Nb, and Ta enrichment of final derivatives. Indeed, taking into account that partition coefficients for these elements in the first phases to crystallize, olivine and diopside, and considerably lower than 1, incompatible elements should be concentrated in late ijolite and Ne-syenite melts. Such distribution is exemplified by Khibiny-type alkaline ultramafics, in which, as follows from the plot in Figure 5, late ijolites are abnormally high in REE, Sr, and Y. On the other hand, as appears from the diagram in Figure 5, within the Kovdor-type alkaline ultramafic series, late ijolites and nepheline syenites of the Maly Kovdor, Vuoriyarvi, and Ozernaya Varaka massifs are more strongly depleted in REE than early differentiates. A similar pattern is found in alkaline ultramafic suites of the Maimechas-Kotui province in Siberia [Egorov, 1991], the Gardiner complex in East Greenland [Nielsen et al., 1997],



Figure 8. Chondrite-normalized REE patterns for rockforming minerals of alkaline ultramafic rocks. Normalizing values, from [Anders and Grevesse, 1989].

and Tanzanian plutonic alkaline suites [Dawson et al., 1995].

Let us consider the main factors responsible for the differences in REE enrichment patterns between Kovdor- and Khibiny-type alkaline ultramafic rocks, which should include (i) conditions at which the primary alkaline ultramafic magma precipitates the principal REE phases (perovskite, apatite, and titanite), and (ii) changes in the composition of the primary magma and, accordingly, in the crystallization

Mineral	Perovskite I					Perovs	kite II			Ap	atite			
Rock	PRX	PRD		PRX		MELT	IJL		PRX			IJ	ΙL	
Massif	AFR	AFR		VUO		KVD	OZV	AFI	R	KVD	OZV	S	LN	KHI
Sample	25-AFR .	AFR-1	$\frac{205}{297}$	282/2224	$\frac{282}{251}$	$\begin{array}{c} 252/\\ 100 \end{array}$	8-OV	25-AFR	AFR-5	7/92.8	8-OV	$\frac{31}{200}$	25/ 110	$rac{1636}{874}$
$\mathbf{Sr}$	894	1796	392	907	545	2168	3545	5846	1274	5005	9358	1997	3202	27612
Ba	4.02	12.5	2.64	2.33	2.58	75.2	8.30	17.5	3.36	21.3	49.7	13.7	22.7	289
Sc	0.28	9.12	4.48	0.04	4.59	n.d.	n.d.	n.d.	4.04	n.d.	n.d.	4.14	4.09	3.97
V	52.1	91.1	34.7	57.7	37.7	173	173	137	39.3	46.8	64.0	26.5	39.7	20.2
Cr	n.d.	12.6	9.82	11.9	13.2	16.1	6.50	n.d.	9.20	n.d.	n.d.	10.3	13.8	9.72
Co	0.66	4.67	0.81	1.15	0.89	n.d.	n.d.	n.d.	0.70	n.d.	n.d.	0.77	0.89	0.75
Ni	1.45	8.67	4.15	5.36	3.56	1.36	0.33	5.01	4.10	4.94	3.27	5.47	8.75	5.63
Cu	80.8	96.4	61.5	72.1	35.1	164	1027	132	23.7	8.17	6.15	17.4	17.3	18.0
Ga	11.2	40.8	14.5	25.6	23.5	78.5	111	9.75	3.22	17.2	22.1	3.24	6.66	14.9
Y	60.0	111	64.4	142	98.6	350	203	131	50.0	149	178	54.9	86.1	134
Nb	991	2915	159	1382	560	4060	8569	3.47	1.34	2.39	13.8	4.42	4.62	9.27
Ta	52.4	234	25.5	91.0	53.7	190	551	0.56	0.06	0.33	3.86	0.12	0.16	0.24
Zr	68.6	213	69.7	186	122	573	142	5.37	0.92	3.11	8.06	0.78	0.47	4.12
Hf	2.25	8.98	2.18	6.40	4.19	16.7	4.54	0.42	0.12	0.38	0.70	0.13	0.15	0.22
Pb	4.64	12.37	4.51	7.12	4.72	30.1	63.8	3.38	1.73	10.4	10.1	2.33	2.83	1.55
U	27.5	34.7	10.3	26.3	19.2	41.5	137	5.61	0.88	22.0	11.3	0.83	3.26	0.74
Th	63.1	456	130	140	129	961	2557	32.5	3.23	339	199	9.24	26.2	4.32
La	1246	3510	880	1962	1629	6587	10700	1573	320	1417	2625	221	515	1798
Ce	3081	9255	2301	4582	4274	14822	22177	2023	499	3068	4423	438	984	2740
Pr	216	803	277	486	460	1645	2210	175	49.1	350	425	49.9	108	239
Nd	683	2570	1039	1716	1621	5644	6719	554	169	1247	1315	184	379	755
Sm	98.1	277	137	236	211	672	677	84.6	25.6	158	151	26.5	51.9	85.5
Eu	25.2	66.9	35.0	63.6	54.6	162	159	25.8	7.86	38.3	40.2	7.74	14.3	23.2
Gd	64.1	178.2	91.7	164	142	328	299	63.9	24.3	89.6	95.9	23.7	42.9	71.4
Tb	7.36	18.8	9.55	18.0	15.4	35.9	31.1	7.93	2.91	9.30	10.8	2.81	4.89	8.22
Dy	25.5	53.2	28.8	66.9	46.2	140	111	32.9	12.0	39.8	45.4	12.0	19.8	32.7
Ho	3.63	6.52	3.92	8.24	6.17	21.1	14.3	5.17	2.02	6.18	7.23	2.19	3.38	5.00
Er	5.91	10.5	6.11	13.7	10.0	36.2	24.0	9.53	4.18	12.0	15.3	4.78	7.12	9.66
Tm	0.64	0.99	0.61	1.33	0.90	3.89	2.60	0.97	0.47	1.49	1.76	0.60	0.81	0.99
Yb	2.80	4.89	2.73	6.38	4.40	19.83	13.92	4.83	1.81	7.16	9.80	2.61	3.80	4.35
Lu	0.28	0.52	0.31	0.69	0.45	2.49	1.88	0.60	0.24	0.78	1.17	0.35	0.46	0.44
$ (La/Yb)_N $	300.3	483.3	217.3	207.3	249.6	223.8	517.9	219.3	119.1	133.4	180.5	57.0	91.4	278.5

**Table 6.** Trace element abundances (ppm) in perovskite and apatite from rocks of the alkaline ultramafic series of the Kola province

order of principal and accessory mineral phases as a result of mixing with batches of phonolitic melt supplied from independent source.

 $\label{eq:constraint} \begin{array}{l} \operatorname{tems}\,SiO_2-TiO_2-Al_2O_3-CaO-MgO-Na_2O \ and \ SiO_2-P_2O_5-Al_2O_3-CaO-MgO-Na_2O. \end{array}$ 

According to experimental data [Kogarko, 1990; Veksler and Teptelev, 1990] and to studies on melt inclusions [Kogarko et al., 1991; Veksler et al., 1998b], perovskite and apatite crystallized at early stages. This is evidenced by the fact that euhedral perovskite and apatite crystals contain inclusions that were homogenized at temperatures of >970°C and 1000–700°C, respectively [Kogarko, 1977; Nielsen et al., 1997]. Hence, crystallization paths of the alkaline ultramafic series must be considered with due account of REEbearing phases, in the context of the six-component sysIn the pseudo-ternary nephelines-diopsides-titanite melting diagram (Figure 10), perovskite and the five remaining phases correspond to the mineral assemblage constituting rocks of the alkaline ultramafic series. According to [Veksler and Teptelev, 1990], perovskite forms a large crystallization field contiguous to the fields of all the phases except olivine. The primary alkaline ultramafic melt of composition A, which contains 2–3 wt % TiO<sub>2</sub>, plots in the diopside crystallization field, so that resultant melts evolve toward the cotectic lines Di–Prv (points B and C) to produce olivine and pyroxene–perovskite cumulates and melil-

Mineral			Titanite		Magnetite			
Rock	PRX		I	JL		DU	N	MELT
Massif	AFR	OZV		KHI		KVD	LSV	KVD
Sample	25-AFR	8-OV	301/500	1636/874	1636/721	4/400	Z-6	252/100
Sr	801	987	4266	4386	2220	n.d.	4.08	56.2
Ba	2.89	16.7	9.69	9.13	6.92	2.38	5.11	13.3
Sc	n.d.	n.d.	n.d.	5.00	4.66	22.7	15.05	n.d.
V	465	273	407	317	127	146	302	735
Cr	0.29	0.86	3.27	17.7	13.7	20300	47780	118
Co	n.d.	n.d.	n.d.	1.11	0.80	238	288	182
Ni	2.34	2.21	2.21	5.51	3.64	3045	1962	796
Cu	227	107	115	61.2	37.0	13.1	31.5	58.6
Ga	7.86	21.3	16.0	11.9	6.92	44.1	35.9	30.8
Y	229	270	334	231	123	1.88	0.22	7.09
Nb	2299	6265	2659	2045	507	5.12	1.64	22.1
Ta	101	593	250	92.5	35.7	0.74	0.23	1.90
Zr	3351	2826	2356	1206	716	46.8	18.3	76.1
Hf	113	73.0	69.0	36.4	17.5	1.54	0.74	2.02
Pb	10.2	10.7	0.15	1.55	24.2	n.d.	0.17	n.d.
U	13.2	13.2	3.12	1.48	1.17	0.01	n.d.	0.01
Th	26.9	100	17.8	9.37	7.71	0.79	0.72	1.50
La	693	1260	961	678	382	8.12	1.22	24.0
Ce	1416	3323	2463	1813	989	12.8	2.25	53.7
Pr	154	407	309	228	123	1.43	0.23	5.69
Nd	556	1423	1148	840	442	4.28	0.57	20.3
Sm	109	209	194	137	70.0	0.60	0.15	2.87
Eu	36.3	59.6	58.7	41.4	20.7	0.01	n.d.	0.84
Gd	98.0	135	143	109	54.7	0.35	0.05	1.93
Tb	13.2	16.9	18.9	14.3	7.29	0.06	0.01	0.27
Dy	62.5	78.6	90.3	63.1	32.2	0.34	n.d.	1.09
Но	10.2	12.8	14.8	10.3	5.33	0.05	n.d.	0.18
Er	20.9	25.7	29.7	20.3	11.1	0.09	n.d.	0.39
Tm	2.40	3.11	3.36	2.25	1.24	0.01	n.d.	0.04
Yb	11.6	16.0	15.4	9.80	5.11	0.03	n.d.	0.18
Lu	1.26	1.87	1.66	0.86	0.49	0.00	n.d.	0.02
$(La/Yb)_N$	40.1	53.2	42.1	46.6	50.4	173.0	_	95.8

Table 7. Trace element abundances (ppm) in titanite and magnetite from rocks of the alkaline ultramafic series of the Kola province

itolites. One factor controlling perovskite stability in melts is silica activity, described by the reactions  $CaTiO_3 + SiO_2$ =  $CaTiSiO_5$  and  $2CaTiO_3 + NaAlSi_3O_8 = NaAlSiO_4 + 2CaTiSiO_5$  [*Carmichael et al.*, 1970; *Veksler and Teptelev*, 1990]. Hence, even small amounts of a more silicic material, when added to silica-undersaturated olivine melanephelinitic melt (the melt evolution course  $A^1 - B^1 - C^1$ ), prevent early crystallization of perovskite. As a result, Ti-silicates will crystallize not in initial melt derivatives in the form of perovskite, but in more evolved products (ijolites and nepheline syenites) in the form of titanite.

The behavior of apatite during evolution of the alkaline ultramafic series can be approximated by the section NaAlSiO<sub>4</sub>-CaMgSi<sub>2</sub>O<sub>6</sub>-Ca<sub>5</sub>(PO<sub>4</sub>)<sub>3</sub>F (Figure 11), as discussed by Kogarko [1990], where the primary olivine melanephelinitic melt falls in the diopside crystallization field. Since the  $P_2O_5$  content of the initial melt is 1.26 wt % (Table 2), apatite is unlikely to precipitate at early stages of rock crystallization. The fact that  $P_2O_5$  distribution in all the members of the series points to the existence of a maximum of 2.8 wt % in melanocratic members of the foidolite trend [Arzamastseva and Arzamastsev, 1996] suggests that apatite appears on the liquidus during crystallization of nepheline–pyroxene assemblages. Based on the above experimental models for crystallization of REE-bearing titanium minerals and phosphates, the following evolution paths for alkaline ultramafic suites of the Kola province can be considered.

Mineral		Olivine			Р	yroxene			Me	lilite	Neph	eline
Rock		DUN		PRX	MELT	IJL	Ν	4LG	M	ELT	IJ	L
Massif	K	VD	LSV	AFR	KV	D	OZV	KHI	K	VD	KVD	KHI
Sample	4/400	4/898	Z-6	25-AFR	252/100	7/30	8-OV	301/500	227/43	252/100	7/30 4	55/402
Rb	4.62	n.d.	n.d.	0.62	0.38	1.27	0.67	2.52	7.03	7.14	117	56.6
Sr	92.5	5.32	1.22	474	301	260	553	2289	3689	3652	64.5	48.7
Ba	4.11	3.55	n.d.	6.25	5.70	5.02	34.9	6.31	7.22	16.5	39.5	57.8
Sc	n.d.	n.d.	n.d.	39.5	51.7	5.57	n.d.	6.17	n.d.	n.d.	9.23	14.2
V	1.40	0.42	0.57	169	41.8	290	131	651	0.76	1.04	0.68	1.60
Cr	3.00	4.55	17.6	4.19	n.d.	138	12.1	0.42	n.d.	n.d.	n.d.	n.d.
Co	77.1	102	151	36.3	13.0	35.4	40.5	34.5	21.3	24.9	n.d.	n.d.
Ni	917	1439	1267	17.9	74.1	42.8	15.2	9.75	4.47	6.55	n.d.	0.70
Cu	3.91	5.15	1.24	216	8.92	6.39	71.0	10.8	3.20	3.23	4.77	2.80
Ga	n.d.	n.d.	n.d.	8.61	2.43	11.44	8.82	6.58	15.8	16.3	45.3	39.4
Y	7.05	1.73	0.51	4.63	5.67	4.19	7.45	17.6	3.54	4.33	2.02	1.68
Nb	7.77	8.27	3.42	8.33	4.50	12.32	10.1	5.44	0.13	0.46	13.0	9.15
Ta	0.80	0.79	0.52	0.51	0.98	0.44	0.95	0.34	0.01	0.08	1.06	0.71
Zr	1.69	2.70	0.50	215	459	452	281	958	0.93	1.17	2.82	7.54
Hf	0.12	0.06	0.06	9.10	12.6	7.18	8.29	24.9	0.03	0.09	0.11	0.20
Pb	1.17	n.d.	0.14	3.65	n.d.	4.65	0.21	1.03	2.66	4.48	0.43	n.d.
Th	3.08	4.32	0.95	0.54	1.69	1.88	1.55	0.75	0.29	0.79	3.92	0.28
La	19.9	17.3	2.29	16.1	14.8	9.17	23.3	36.6	39.2	38.3	22.40	4.81
Ce	33.0	29.8	3.85	29.1	32.4	22.6	52.5	84.9	70.2	66.2	40.3	9.42
Pr	3.34	3.21	0.42	3.09	3.87	3.19	6.04	10.0	7.48	7.01	3.68	1.00
Nd	11.5	9.44	1.31	11.1	15.4	12.1	22.4	38.0	25.2	25.2	12.6	3.35
Sm	1.87	1.20	0.07	1.85	2.73	1.94	3.46	6.45	3.48	3.40	1.72	0.64
Eu	0.33	0.19	0.002	0.53	0.68	0.35	0.94	1.95	0.67	0.78	0.30	0.03
Gd	1.19	0.73	0.10	1.76	1.82	1.39	2.51	4.88	1.82	1.81	1.04	0.57
Tb	0.18	0.08	0.02	0.25	0.23	0.14	0.34	0.66	0.19	0.20	0.12	0.07
Dy	0.98	0.31	0.06	1.15	1.15	0.74	1.81	3.67	0.95	0.93	0.45	0.34
Ho	0.19	0.02	0.01	0.16	0.19	0.14	0.30	0.71	0.15	0.13	0.05	0.05
Er	0.48	0.02	n.d.	0.38	0.46	0.36	0.83	1.99	0.24	0.25	0.08	0.10
Tm Nu	0.07	0.001	n.d.	0.06	0.07	0.07	0.13	0.36	0.03	0.03	0.01	0.01
Yb	0.51	n.d.	n.d.	0.58	0.52	0.95	1.00	3.39	0.16	0.13	0.03	0.07
	0.09	n.d.	n.d.	0.10	0.09	0.19	0.18	0.66		0.02		0.01
$(La/Yb)_N$	27.7	_	_	19.7	20.2	6.8	16.4	7.6	177.8	202.8	463.3	47.7

Table 8. Trace element abundances (ppm) in rock-forming minerals of the alkaline ultramafic series of the Kola province

# 6.1. Evolution of the Kovdor-type alkaline ultramafic series

liquidus as well. With respect to the Kovdor-type alkaline ultramafic series, evidence for the model just proposed is as follows: (1) The Afrikanda, Vuoriyarvi, Salmagora, and Cape

According to the first scenario, which is apparently materialized in the Kovdor-type rock sequence, the olivine melanephelinitic melt A of Figure 10, after having precipitated olivine, will evolve toward the diopside-perovskite cotectic, where perovskite-clinopyroxene cumulates are formed. Early crystallization of a phase that has  $D_{\text{REE}} \gg 1$ will result in a dramatic depletion of residual melt, which will sequentially produce the series of REE-depleted derivatives (ijolites and nepheline syenites). The central role in early extraction of REE from melt was thus played by perovskite, although apatite may also have played a subordinate role at this stage, because it could, in principle, have reached the (1) The Afrikanda, Vuoriyarvi, Salmagora, and Cape Turiy massifs contain clinopyroxene–perovskite cumulates composed of primary magmatic perovskite. In some intrusions (Kovdor, Afrikanda, Ozernaya Varaka, Vuoriyarvi), local clinopyroxenite zones with up to 3% primary magmatic apatite have been found.

(2) The latest derivatives of the Kovdor series are sharply depleted in REE, Nb, Ta, and Sr relative to both the mean parental magma composition and to earlier cumulates.

(3) Primary REE-bearing accessories in late differentiates of the Kovdor series are relatively low in REE and Sr due to the melts being depleted in these elements. Thus, apatites from Ozernaya Varaka cancrinite symplex have as little

	$\mathrm{D}_{\mathrm{Per}/\mathrm{Ap}}$	$\mathrm{D}_{\mathrm{Per}/\mathrm{Ap}}$	$\mathrm{D}_{\mathrm{Per}/\mathrm{Ap}}$	$\mathrm{D}_{\mathrm{Per/Tit}}$	$\mathrm{D}_{\mathrm{Per/Tit}}$	$\mathrm{D}_{\mathrm{Ap/Tit}}$	$\mathrm{D}_{\mathrm{Ap/Tit}}$	$\mathrm{D}_{\mathrm{Ap/Tit}}$
Rock	PRD	PRX	MLG	PRX	IJL	PRX	MLG	IJL
Sample	AFR-5	25-AFR	8-OV	25-AFR	8-OV	25-AFR	8-OV	1636/874
La	1.43	0.79	4.08	1.80	8.49	2.27	2.08	2.65
Ce	1.79	1.52	5.01	2.18	6.67	1.43	1.33	1.51
$\Pr$	1.86	1.23	5.20	1.40	5.42	1.14	1.04	1.05
Nd	1.89	1.23	5.11	1.23	4.72	1.00	0.92	0.90
$\operatorname{Sm}$	2.30	1.16	4.49	0.90	3.24	0.78	0.72	0.63
Eu	2.73	0.98	3.94	0.69	2.66	0.71	0.67	0.56
Gd	2.61	1.00	3.12	0.65	2.21	0.65	0.71	0.66
$^{\mathrm{Tb}}$	3.32	0.93	2.87	0.56	1.84	0.60	0.64	0.57
Dy	4.38	0.77	2.43	0.41	1.41	0.53	0.58	0.52
Ho	4.99	0.70	1.97	0.35	1.11	0.51	0.57	0.49
Er	5.67	0.62	1.57	0.28	0.93	0.46	0.59	0.48
Tm	6.58	0.66	1.48	0.27	0.84	0.40	0.57	0.44
Yb	8.77	0.58	1.42	0.24	0.87	0.42	0.61	0.44
Lu	7.69	0.46	1.61	0.22	1.01	0.48	0.63	0.51

 Table 9. Partition coefficients between the coexisting perovskite, apatite, and titanite in the rocks of the alkaline ultramafic series

as 0.67-0.71 LREE<sub>2</sub>O<sub>3</sub> and 0.60-0.68 wt % SrO, and those from Maly Kovdor nepheline syenites, as little as 0.33-1.31 LREE<sub>2</sub>O<sub>3</sub> and 0.59-1.07 wt % SrO [Arzamastseva and Arzamastsev, 1996].

Data on REE distribution in alkaline intrusions of various provinces indicate that early fractionation of REE-bearing phases is a characteristic feature of alkaline ultramafic suites. Thus, the Oldoinyo Lengai plutonic suite has been reported to comprise clinopyroxenite (jacupirangite) cumulates with up to 28% perovskite [*Dawson et al.*, 1995]. The character of REE distribution in the Oldoinyo Lengai rocks corresponds exactly to that in the Kovdor series: REE are enriched in early cumulates and sharply depleted in terminal members



Figure 9. Whole rock/perovskite partition coefficients for alkaline ultramafic rocks, using data from Table 10. Coefficients for perovskite in alkaline volcanites from other regions (analysis 1, heavy line), from [Onuma et al., 1981].



of the series (ijolites and eucolite-bearing Ne-syenites). An-

other example is provided by the Maimecha-Kotui province,

where the Kugda, Guli, and Odikhincha massifs are re-

ported to contain olivine and clinopyroxene cumulates rich

Figure 10. Melting diagram for the system nephelinesdiopsides-titanite, after [Veksler and Teptelev, 1990]. Di – diopside, Prv – perovskite, Ttn – titanite, Mel – melilite, Cg – carnegieite, Ne – nepheline, Ol – olivine, L1 + L2 – immiscibility region. Point A signifies the weighted mean composition for the Kovdor-type primary alkaline ultramafic magma. Point  $A^1$  represents the hypothetical primary magma of the Khibiny-type alkaline ultramafic series, whose composition has been changed by mixing with phonolitic melt.

			Perovskite	;		Apa	tite
Rock	PRX	PRD	PRX	PRX	PRX	PRX	PRX
Sample	25-AFR	AFR-1	282/224	205/297.5	282/251	25-AFR	AFR-5
La	9.5	26.7	14.9	6.7	12.4	12.0	2.4
Ce	12.4	37.1	18.4	9.2	17.1	8.1	2.0
Pr	7.9	29.3	17.7	10.1	16.8	6.4	1.8
Nd	7.0	26.5	17.7	10.7	16.7	5.7	1.7
Sm	6.8	19.3	16.4	9.5	14.7	5.9	1.8
Eu	7.0	18.6	17.6	9.7	15.1	7.2	2.2
Gd	6.3	17.4	16.0	9.0	13.9	6.2	2.4
Tb	5.8	14.9	14.3	7.6	12.2	6.3	2.3
Dy	4.2	8.8	11.1	4.8	7.7	5.5	2.0
Ho	3.5	6.2	7.9	3.7	5.9	4.9	1.9
Er	2.4	4.3	5.7	2.5	4.1	3.9	1.7
Tm	2.0	3.1	4.2	1.9	2.8	3.1	1.5
Yb	1.5	2.7	3.5	1.5	2.4	2.6	1.0
Lu	1.1	2.0	2.7	1.2	1.7	2.3	0.9

Table 10. Mineral/rock partition coefficients in pyroxene and olivine cumulates of the alkaline ultramafic series

Notes.  $D_{Prv/Rock}$  and  $D_{Ap/Rock}$  were calculated with respect to the weighted mean composition of the alkaline ultramafic series (see AVER<sup>\*</sup> in Table 4).

in primary magmatic perovskite [Egorov, 1991]. The Nizhnesayansky carbonatite massif (southern Siberia) has also been reported to incorporate a rock suite encompassing the full spectrum of alkaline ultramafites, including clinopyroxene cumulates with as much as 15% perovskite [*Chernysheva et al.*, 1990].



Figure 11. Melting diagram for the system nephelinesdiopsides-apatite, after [Kogarko, 1990]. Di – diopside, Ap – apatite, Pho – Ca-phosphate, Mel – melilite, Cg – carnegieite, Ne – nepheline, Ol – olivine, L1 + L2 – immiscibility region. Point A signifies the weighted mean composition for the Kovdor-type primary alkaline ultramafic magma, and point  $A^1$ , the hypothetical composition for the primary magma of the Khibiny-type alkaline ultramafic series.

# 6.2. Evolution of the Khibiny-type alkaline-ultramafic series

The set of rock varieties that make up fragments of the alkaline ultramafic suite in the Khibiny massif, the order of their formation, and overall major-element characteristics were formerly believed to suggest similar courses of REE evolution and distribution for the Khibiny and Kovdor suites. However, according to geologic observations and geochemical data for the Khibiny alkaline ultramafic suite, generation of early olivine and clinopyroxene cumulates was not accompanied by massive precipitation of perovskite. This is the reason why REE remained in residual melt until final stages of crystallization, and REE enrichment did not occur until late ijolite derivatives started to form. Two factors preventing early crystallization of perovskite can be considered, both of them being apparently related to a drastic increase in silica activity in the melt. Firstly, the change in SiO<sub>2</sub> activity may result from interaction of the primary olivine melanephelinitic magma with surrounding Precambrian basement rocks. This, however, is at variance with Sr and Nd isotope data, which suggest that crustal material was not involved in alkaline magmagenesis in the Kola province [Kramm and Kogarko, 1994]. Secondly, the change in  $SiO_2$  activity may have been caused by mixing of the olivine melanephelinitic magma with a more silicic melt. A likely candidate for such a melt might be the phonolitic magma that gave rise to the Khibiny agpaitic plutonic suite. This suite, according to isotope geochemical and petrologic

Table 11. Mineral/rock partition coefficients in melilitic rocks and ijolites of the alkaline ultramafic series

Apatite						Titani	ite		Clir	opyroxer	ne	Mel	ilite	
Massif	SLN	SLN	KVD	OZV	KHI	OZV	KHI	KHI	OZV	KVD	KVD	KHI	KVD	KVD
Rock	IJL	IJL	IJL	MLG	IJL	MLG	IJL	IJL	MLG	IJL	MELT	IJL	MELT	MELT
Sample	25/110	31/200	7/30	8-OV	1636/721	8-OV	301/500	1636/721	8-OV	7/30	252/100	301/500	252/100	227/43
La	4.56	5.72	25.57	15.54	8.81	7.46	8.83	1.87	0.14	0.17	0.35	0.34	0.90	1.37
Ce	4.55	5.27	25.96	14.04	8.26	10.55	10.70	2.98	0.17	0.19	0.43	0.37	0.87	1.33
Pr	4.55	5.06	26.64	12.45	7.90	11.94	12.32	4.06	0.18	0.24	0.48	0.40	0.87	1.31
Nd	4.39	4.81	27.11	11.34	7.86	12.27	12.85	4.60	0.19	0.26	0.56	0.43	0.92	1.26
Sm	4.08	3.99	26.51	9.97	7.13	13.83	13.64	5.82	0.23	0.33	0.73	0.45	0.91	1.29
Eu	3.94	3.86	26.39	11.18	7.16	16.56	14.83	6.39	0.26	0.24	0.82	0.49	0.93	1.28
Gd	3.94	3.94	25.81	8.64	7.62	12.21	12.70	5.84	0.23	0.40	0.88	0.43	0.88	1.12
Tb	3.52	3.24	25.27	8.47	6.85	13.22	12.80	6.08	0.26	0.38	1.06	0.45	0.91	1.15
Dy	3.28	2.48	26.89	8.64	6.69	14.94	13.82	6.59	0.34	0.50	1.27	0.56	1.03	1.34
Ho	3.04	2.61	23.32	8.82	5.82	15.61	12.96	6.20	0.36	0.53	1.39	0.63	0.97	1.21
Er	2.84	2.35	16.89	8.18	4.91	13.77	12.20	5.64	0.44	0.51	1.44	0.82	0.80	0.88
Tm	2.50	2.03	14.50	7.04	3.44	12.43	11.68	4.32	0.51	0.67	1.67	1.23	0.65	0.76
Yb	2.22	1.47	9.97	6.53	2.23	10.64	10.07	2.62	0.66	1.32	2.22	2.22	0.57	0.63
Lu	2.02	1.18	6.29	4.67	1.28	7.47	9.61	1.44	0.72	1.56	3.38	3.80	0.69	0.53

Notes.  $D_{Ap/Rock}$ ,  $D_{Ttn/Rock}$ ,  $D_{Cpx/Rock}$  and  $D_{Mel/Rock}$  were calculated from the ratios of REE abundances in rocks and mineral phases contained in them.

data [Arzamastsev et al., 1998; Kramm et al., 1993], evolved independently, and its origin is related to a mantle source other than that of alkaline ultramafic melts. The main feature of agpaitic melts is their relatively higher  $SiO_2$  (52–56 wt %), alkali (Na<sub>2</sub>O +  $K_2O > 16$  wt %), and F contents. However, REE contents in agpaitic syenites, according to our data [Arzamastsev et al., 2001a], are only a little higher than in alkaline ultramafites. Hence, addition to the primary silica-undersaturated melanephelinitic magma of even minor batches of phonolitic melt would increase silica and alkali contents of the magma while changing its REE content only slightly. Indeed, in comparison to Vuoriyarvi, Ozernaya Varaka, Sallanlatva, and Kovdor ijolites, their Khibiny counterparts are much higher in SiO<sub>2</sub>, Na<sub>2</sub>O, and K<sub>2</sub>O (Tables 2, 3), and feldspar-bearing varieties are widespread among them. In the nepheline-diopside-titanite melting diagram (Figure 10), the change in the composition of crystallizing melt will be expressed in that the initial liquid of composition A shifts away from the perovskite cotectic surface  $A^1$  to produce perovskite-free olivine-diopside, diopside-melilite, and diopside-nepheline rocks (option  $A^1 - B^1 - C^1$ ). Accordingly, REE will be enriched progressively in these differentiates, and extraction of REE from melt will not occur until the melt has reached the apatite and/or titanite liquidus surface at the final stages of alkaline ultramafic petrogenesis. Therefore, evolution of the alkaline ultramafic series in the Khibiny massif was disturbed by mixing of minor portions of phonolitic melt with the primary olivine melanephelinitic magma, which led to a change in crystallization order of REE-bearing titanates and Ti-silicates and enrichment of late batches of melt in the majority of incompatible elements.

#### 7. Conclusions

Our study of REE distribution in rocks and minerals of Paleozoic alkaline ultramafic rocks of the Kola province affords the following conclusions:

(1) REE patterns in rocks of the Kovdor, Afrikanda, Vuoriyarvi, and Salmagora massifs indicate a systematic depletion from earlier to later (ijolite and nepheline syenite) melt derivatives. Analysis of rock suites to be found in alkaline intrusions of other provinces (Maimecha–Kotui province of southern Siberia and East African province) shows that the above trend has a general character, inherent in many alkaline ultramafic suites.

(2) REE distribution in Kovdor-type alkaline ultramafic suites is controlled by crystallization of perovskite and, to a lesser extent, apatite. Primary olivine melanephelinitic melts of this series experienced crystallization of perovskite, the main REE-bearing mineral. Perovskite coprecipitating with the first phases to crystallize from melt, olivine and clinopyroxene, leads to a dramatic REE depletion of the residual melt and to formation of REE-depleted derivatives, ijolite and nepheline syenite.

(3) Petrogenesis of the alkaline ultramafic suite of the Khibiny massif was upset by mixing of minor batches of phonolitic melt with the primary olivine melanephelinitic magma, with an ensuing change in the crystallization order of REE-bearing titanates and Ti-silicates and enrichment of late batches of melt in the majority of incompatible elements. As a result, Khibiny ijolites, which are late and the most evolved derivatives of the alkaline ultramafic

magma, have the highest REE concentrations, accommodated by high-REE apatite and titanite.

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