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# 2.9 to 1.9 Ga paleoalterations of Archean granitic basement of the Franceville basin (Gabon).

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- 4

#### Abstract

The Archean granitoids in the Kiéné area, Gabon, are overlained by the Paleoproterozoic sediments of the Franceville basin (2.1 Ga). The basin is known for its high-grade uranium deposits among which some have been forming natural nuclear fission reactors. Most of the studies were dedicated to the FA-FB Paleoproterozoic sediments hosting these uranium deposits. Little is known on the Archean basement itself and specifically on the hydrous alteration events it experienced before and after the sediment deposition. The present work is focused on their petrographical, mineralogical and geochemical characterization. Dating the successive alteration events has been attempted on altered monazite crystals.

Rocks in different alteration states have been sampled from eight drill cores crosscutting the Archean - Paleoproterozoic unconformity. The Archean granitoids observed in the deepest levels exhibit typical petrographical features of a propylitic alteration while they are intensely illitized up to the unconformity. The propylitic alteration is mainly pervasive but the original texture of the granitoïds is conserved in spite of the formation of new minerals: Mg-chlorite, allanite and epidote forming a typical paragenesis. The illitic alteration is much more invasive near the unconformity. The illitization process leads to the replacement of feldspars and the corrosion of quartz crysals by an illitic matrix while the ferromagnesian minerals are pseudomorphosed by a Fe-chlorite + phengite+ hematite assemblage. The final fluid-rock interaction step is marked by fissural deposits of calcite and anhydrite. The  $\delta^{13}$ C isotopic data show that the fissural carbonates precipitated from diagenetic fluids enriched carbon products deriving from the maturation of organic matter. The U-Pb isotopic analyses performed on monazite crystals have dated three distinct events: 3.0-2.9 Ga (magmatic), 2.6 Ga (propylitic alteration) and 1.9 Ga (diagenetic illitization). The calculation of geochemical mass balances suggests that the water - rock ratio during the propylitic alteration event was weak. On the contrary, it was much higher during the overprinted illitization which is characterized by an intense leaching of Na, Ca, Mg, Sr, REE and an enrichment in K, Rb,Cs.

Neither the petrographic features nor the geochemical data militate for an Archean weathering event (paleosol). In the present case, diagenetic fluids have percolated from the unconformity into the basement where they overprinted the illitization processes upon the previously propylitized rocks. These fluids were probably oxidant as they are also responsible of the U mobilization which led to the formation of the ore deposits close to the FA-FB interface.

5 Keywords: Archean, granitoid, hydrothermal, Paleoproterozoic, diagenesis, Francevillian

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9 Introduction

The Archean basements have been frequently altered by multiple hydrothermal and/or 10 metamorphic events (Card, 1978; Murakami et al., 2011; Nedachi et al., 2005; Ohmoto, 1996; 11 Ohmoto and Kerrick, 1977). Additionally, in some cases, their erosion surface may conserve the 12 traces of weathering processes forming the so-called paleosols which are researched for their 13 paleo-atmospheric signification (Grandstaff et al., 1986; Holland, 1994; Maynard, 1992; 14 Mossman and Farrow, 1992; Ohmoto, 1996; Rye and Holland, 1998). At last, when overlain by 15 sedimentary deposits, the basement rocks may have suffered mineral reactions triggered by the 16 diagenetic fluid invasion (Mercadier et al., 2010). Because they are operating successively, all 17 these events may superimpose their effects in some places. Then, the genetic message carried by 18 19 the secondary minerals is difficult to interpret without an accurate petrographical study. This is the backbone of the present work whose goal is to characterize the alteration events which have 20 affected the Archean basement under the Francevillian sedimentary series in Gabon. It was 21 globally affected by post-magmatic to metamorphic mineral reactions (Mathieu, 1999; Sère, 22 1996) but its alteration state near the unconformity is still poorly documented. Our work, focused 23 24 on that interface, was made possible thanks to several drill-holes which have been dug for 25 uranium ore exploration. The goal is to determine the origin of the secondary minerals and to date their formation. 26

#### 28 2. Geological setting

The Chaillu massif is located in the South Gabon which is Northeastern part of the Congo 29 30 craton (Fig.1). It forms the granitic basement of the Paleoproterozoic sediments deposited in the Franceville basin. It is composed of different granitoid and orthogneiss bodies dated from 3.1 to 31 2.5 Ga (Bouton et al., 2009; Caen-Vachette et al., 1988). These bodies are structurally and 32 33 mineralogically heterogeneous because of two major plutonic events (Bouton et al., 2009; Chevallier et al., 2002; Prian and Johan, 1989; Thiéblemont et al., 2009; Thomas et al., 2001). 34 The first one took place during the Mesoarchean (2928  $\pm$  6 Ma to 2870  $\pm$  5 Ma). It led to the 35 intrusion of gray-coloured calc-alkaline granitoids belonging to typical a Archean Trondhjemite, 36 Tonalite, Granodiorite suite (TTG) associated with migmatized quartz diorites (Thiéblemont et 37 al., 2009). Locally, these rocks are cross-cut by diorite sills. The second period of magmatism 38 belongs to the Neoarchean (2800 to 2550 Ma). It is characterized by the intrusion of alkaline to 39 calc-alkaline pink to red magmatic rocks: syenites, syenogranites, granites and pegmatites 40 41 (Thiéblemont et al., 2009).

The Franceville basin sedimentary series are composed of five unmetamorphosed formations, 42 4000 m thick, sequentially labeled from FA at the base up to FE (Weber, 1968). The FA 43 formation (500-1000 m thick) is directly deposited on the eroded surface of the Archean 44 granitoids basement (Chaillu massif). It is essentially formed of conglomerates and sandstones 45 typical of fluvio-deltaic deposits. Its upper part (Fig. 1) hosts the natural nuclear reactors of the 46 Oklo, Okélobondo and Bangombé uranium ore bodies (Gauthier-Lafaye et al., 1996; Naudet, 47 1991). The FA formation in the Kiéné area, exhibits a vertical granulometric evolution 48 (conglomerate to sandstone and mudstone). The color of the sediments changes upwards from red 49 50 to green to black (Gauthier-Lafaye, 1986; Haubensack, 1981). The fine-grained black levels are

composed of detrital quartz, plagioclase and biotite. Plagioclase and biotite are locally replaced 51 52 by calcite and illite + chlorite assemblage respectively (Haubensack, 1981). The green colored intermediate levels are rich in illite (or phengite) and chlorite which replace detrital biotite and 53 locally feldspar. The sandstone is mainly composed of quartz grains cemented by calcite, illite 54 and chlorite. The conglomeratic sandstones at the base of the FA sedimentary series lie directly 55 on the granitoid basement. Their mineralogical composition is dominated by a microcrystalline 56 illitic material filling the porosity and replacing feldspars. When disseminated between quartz 57 crystals, the illitic matrix may contain calcite, anhydrite or gypsum. The sandstones at the very 58 base of the series, are known to be red colored due to the precipitation of hematite in the 59 cleavages of altered biotites and to the staining of detrital quartz grain surfaces. This is not the 60 case in the eight studied drills crosscutting the unconformity we studied here: no trace of red 61 coloration has been observed probably because detrital ferromagnesian minerals are lacking or 62 perhaps, because of a secondary discoloration. 63

64 **3. Sampling and analytical methods** 

#### 65 3.1 Sampling

The eight drill cores studied here are located in the Kiéné region (250 km<sup>2</sup>), the central part of the Franceville basin, close to the Mikouloungou U-ore deposit and the Kaya-Kaya fault (Fig. 2a). Petrographic investigations were performed on 38 samples collected along the eight drill cores (KA 13, GR1, GR5, GR23, GR31, GR20 GR43 and KA6) down to variable depth (5 to 36 m) below the unconformity between sandstones and granitic basement (Fig. 2b). From the deepest levels to the sediment interface, the granitoids exhibit three dominant "facies" which correspond to an increasing alteration degree: *(i)* the least altered one, observed at depth, are gray

TTG or pink-to-red syenitic-to-granitic rocks; *(ii)* the altered granitoids in which the initial petrographical texture is preserved; *(iii)* the intensely altered granitoids which are green colored close to the sandstone-basement unconformity for all the studied drill-cores. The granitoids are largely altered and red colored near the sandstone contact in the GR20 and GR31 drill cores (Fig. 2b).

#### 78 **3.2 Analytical methods**

The petrographical observations of the granitoids were performed using optical and scanning 79 electron microscopes (OM, SEM). Polished thin sections were carbon coated before examination 80 using the backscattered electron mode imaging (Jeol JSM 5600LV). X-ray spectra were collected 81 with an energy dispersive device according to following analytical conditions: 15 kV accelerating 82 voltage, 1 nA probe current and a 16.5 mm focalization distance. Micro-chemical analyses were 83 84 obtained using a CAMECA SX50 electron microprobe (CAMPARIS analytical center, University Pierre and Marie Curie, Paris VI) equipped with wavelength dispersive spectrometers (15 kV 85 accelerating voltage, 1 nA probe current, 1µm beam spot diameter). Bulk rock analyses were 86 87 carried out at the SARM-CRPG (Nancy, France; Table 1). The samples were fused with LiBO<sub>2</sub>, dissolved with HNO<sub>3</sub>, analyzed using ICP-EAS and ICP-MS for major and trace elements 88 89 respectively. The calibration is based on international standards. The mineralogical composition of each sample has been determined using X-ray diffraction of randomly oriented powders 90 91 (Bruker D8 ADVANCE A25 diffractometer,  $CuK\alpha$ ). The accelerating voltage and probe current 92 were of 40KV and 40 mA, respectively. Clay mineral identification was detailed using oriented <  $2\mu$ m preparations scanned in the air dried and ethylene glycol saturation states in the 8 to  $30^{\circ} 2\theta$ 93 CuKα angular range. The C and O isotope compositions of calcitic veins were measured in the 94 95 stable isotope laboratory of the Géosciences department (Rennes, University of Rennes 1).

96 Calcite was reacted with anhydrous H<sub>3</sub>PO<sub>4</sub> at 50°C and the isotopic compositions measured on a
97 triple collector VG Optima mass spectrometer.

98 The monazite crystals observed in two samples (KA6 437.85 and GR1 631) near the unconformity level were used to constrain the ages of plutonic rock emplacement and their 99 subsequent alterations. These crystals were examined using SEM, mostly in backscattered 100 101 electron mode, and analyzed using a CAMECA SX50 electron microprobe (CAMPARIS center, University Pierre and Marie Curie, Paris VI). U-Th-Pb isotopic data were obtained by laser 102 ablation inductively coupled plasma spectrometry (LA-ICPMS) at the Volcanology Laboratory of 103 Clermont-Ferrand University. The analyses involved the ablation of minerals using a Resonetics 104 Resolution M-50 powered by an ultra-short-pulse (<4ns) ATL Atlex Excimer laser system 105 operating at a wavelength of 193 nm (Muller et al., 2009). A 7 µm laser spot diameter and 106 repetition rate of 1 Hz with energy of 8 mJ producing a fluence of 15 J/cm<sup>2</sup> was used for zircon 107 dating. The ablated material was carried into helium flux and then mixed with nitrogen and argon 108 before injection into the plasma source of an Agilent 7500 cs ICP-MS equipped with a dual 109 pumping system to enhance the sensitivity. The alignment of the instrument and mass calibration 110 were performed before each analytical session using the NIST SRM 612 reference glass, by 111 inspecting the signal of  $^{238}$ U and by minimising the ThO<sup>+</sup>/Th<sup>+</sup> ratio (< 1%). The mean sensitivity 112 for <sup>238</sup>U using a spot size of 44 µm is about 25,000 cps/ppm. The analytical method for isotope 113 dating of zircon with laser ablation ICPMS is similar to that developed for monazite and reported 114 by Paquette and Tiepolo (2007). The signals of <sup>204</sup>(Pb+Hg), <sup>206</sup>Pb, <sup>207</sup>Pb, <sup>208</sup>Pb, <sup>232</sup>Th and <sup>238</sup>U 115 masses are acquired. The occurrence of common Pb in the sample can be monitored by the 116 evolution of the <sup>204</sup>(Pb+Hg) signal intensity, but no common Pb correction was applied owing to 117 the large isobaric interference from Hg. The <sup>235</sup>U signal is calculated from <sup>238</sup>U on the basis of the 118

ratio  ${}^{238}U/{}^{235}U= 137.88$ . The analyse procedure is based on 30 seconds of background integration 119 with laser off followed by 1 minute integration with the laser firing and a 30 seconds delay to 120 wash out the previous sample and prepare the next analysis. Data are corrected for U-Pb 121 fractionation occurring during laser sampling and for instrumental mass discrimination (mass 122 123 bias) by standard bracketing with repeated measurements of the Moacyr monazite standard (Gasquet et al., 2010). Repeated analyses of Manangoutry monazite standard (Paquette and 124 Tiepolo, 2007), treated as unknown, independently control the reproducibility and accuracy of the 125 corrections. Data reduction was carried out with the software package GLITTER<sup>®</sup> from 126 Macquarie Research Ltd (Achterbergh et al., 2001; Jackson et al., 2004). For each analysis, the 127 time resolved signal of single isotopes and isotope ratios was monitored and carefully inspected 128 to check the presence of perturbations related to inclusions, fractures, mixing of different age 129 domains or common Pb. Calculated ratios were exported and Concordia <sup>206</sup>Pb/<sup>238</sup>U vs <sup>208</sup>Pb/<sup>232</sup>Th 130 ages and diagrams were generated using the Isoplot/Ex v. 3.23 software package (Ludwig, 131 2001). The concentrations in U-Th-Pb were calibrated relative to the certified contents of Moacyr 132 monazite standard (Seydoux-Guillaume et al., 2002). 133

134 **4. Results** 

#### 135 **4.1. The least-altered granitoids**

The least altered gray or pink-to-red granitoids are observed in the deepest parts of the different drill cores far from the unconformity (Fig. 2b). The primary minerals are quartz, plagioclase ( $An_0 - An_{20}$ ), biotite, alkali feldspar and locally amphibole. Zircon, apatite, uranothorite and monazite are the common accessory mineral phases. The gray granitoids present sometimes a gneissic structure.

The magmatic rocks range from diorite to granite. The whole rock major element compositions 141 142 of the less altered rocks reflect the variety of these rock types (Table 1): i.e.  $SiO_2$  (52.1 - 74.5 wt%), FeO<sub>T</sub> (0.3 - 9.5 wt%), MgO (0.16 - 6.7 wt%). In order to determine their magmatic type, 143 the chemical composition of these rocks has been plotted in the Q-P chemical-mineralogical 144 diagram of Debon and Le fort (1983) (Fig.3). Because even the best preserved samples from the 145 eight drill cores exhibit petrographic features of a hydrothermal alteration event the rock samples 146 were selected according to their low ignition loss value: 2.5 Wt%. The grey TTGs range from 147 quartz-diorite (dq) to tonalite (to) composition, while the pink and red rocks correspond to 148 granodiorite (gd) and adamellite (ad) (Table 1; Fig. 3). The chondrite-normalized REE patterns 149 show that tonalities present the typical highly fractionated pattern: high LREE content, low 150 HREE content, no significant Eu anomaly. These properties are typical of the Archean TTG 151 (Martin, 1986). On the other hand, the granodiorite and adamellite have lower REE contents with 152 a positive Eu anomaly; such REE patterns have already been interpreted in Archean rocks as 153 resulting from the fractionation by accessory phases such as allanite and zircon (Martin, 154 1987)(Fig. 4). Finally the higher REE content of the diorite is similar to that of the late Archean 155 156 sanukitoids (Martin et al., 2010; Stern, 1989). In other words, all the less altered rock samples display the features typical of late Archean magmatic associations. 157

#### 158 **4.2. The altered granitoids**

#### 159 4.2.1. Samples located far from the unconformity

All the Kiéné granitoid samples located far from the unconformity exhibit the presence of chlorite. A pervasive chloritization process has severely affected the primary ferromagnesian minerals, i.e. total replacement of biotite and amphibole. Chlorite is associated to epidote, titanite

and allanite (Fig. 5a). According to Lowell and Guilbert, (1970) and Titley et al. (1986), this 163 164 paragenesis is typical of a propylitic alteration. Concomitantly, the plagioclases are partially replaced by small sized micas (sericite) forming a microcrystalline matrix in which calcite, pyrite 165 and chalcopryrite crystals are locally disseminated. Some fractures crosscut the propylitzed 166 rocks. They are filled by a zoned deposit: chlorite along the walls and calcite sealing the central 167 JSCR part (fig. 5b) 168

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#### 4.2.2. Samples close to the unconformity 170

The Kiéné granitoids are affected by an intense illitization in the vicinity of unconformity 171 giving them a green color. This alteration is mainly invasive and extends down to 4-5 m depth. 172 Plagioclase, K-feldspar and previously chloritized biotite are totally or partially replaced by illite 173 (Fig.6a and b). The feldspars are corroded by a microcrystalline matrix (5-20 µm) while the 174 chloritized biotite are replaced by larger illite crystals (10-50µm). The quartz bodies, themselves, 175 exhibit corrosion features and are locally crosscut by illite veinlets. The illitization intensity 176 decreases with depth becoming nearly undetectable between 5 and 7 m down from the 177 178 unconformity. Here, only illite veinlets are still observable. Close to the unconformity, the GR20, GR23 and GR31 granitoid samples become red colored. The chloritized biotites are replaced by 179 180 an illite + hematite assemblage (Fig. 6b). The rocks are crosscut by calcite ± anhydrite veins (Fig. 181 6c). Some monazites have been locally observed disseminated through the illitic matrix.

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#### 4.2.3. Alteration features of FA sandstones 183

The FA sandstones above the unconformity exhibit detrital quartz and K-feldspar crystals embedded in an illitic micro-crystalline matrix while the detrital grains, whatever their composition, are invaded along their inter-granular joints. Samples GR23 629 and GR20 818 are cemented by anhydrite and/or calcite deposits (Fig. 6d).

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#### 189 4.3. Geochemical data of bulk rocks

Because of the predominance of hydrolyzed and carbonated secondary phases in the secondary 190 minerals, the loss on ignition (L.O.I) has been chosen here as an indicator of the alteration 191 degree. In most samples, L.O.I. is significantly greater than that of the corresponding unaltered 192 rocks. Indeed, the measured values range from 0.76 wt% in the freshest granitoids to 4.5 wt% for 193 the most altered. However, the average L.O.I. calculated from a compilation of 60 TTG samples, 194 is much lower: 0.25 wt% (Huang et al., 2013). Consequently, the absolute value of L.O.I. cannot 195 be directly used as a reliable marker because of two reasons: (i) for a given rock type, it depends 196 on the quantity of secondary minerals produced by alteration processes; (ii) it varies according to 197 198 rock types. For instance, diorite L.O.I. typically averages to 1.55 wt% (Debon and Le Fort, 1983), 199 tonalite for 0.32 and granodiorite for 0.12 (Huang et al., 2013). The tonalite value will be used here as the reference in the calculation of relative L.O.I. variations  $\Delta LOI = (LOI_{sample} -$ 200  $LOI_{ref}$ /(LOI<sub>most altered sample</sub> – LOI<sub>ref</sub>). According to this definition, the  $\Delta LOI$  calculated for each 201 202 basement sample shows a regular increase toward the unconformity (Fig. 9). The most altered 203 levels are found in the illitic zone in agreement with petrographic observations. In order to address the problem of element mobility, it appears suitable to normalize all chemical element 204 values to that one remaining immobile during the alteration processes. As already proposed by 205

several authors (Nesbitt, 1979; Ohmoto, 1996; Rye and Holland, 1998; White et al., 2001), Ti or 206 207 Zr are classically considered as reference immobile elements. Here, Ti has been chosen because of its higher content. Then, each element content in the altered samples is corrected by the  $TiO_2$ 208 reference/TiO<sub>2 sample</sub> ratio. Propylitic and illitic processes being acceptably discriminated by the 209 210 increasing loss on ignition (Fig. 7), the  $\Delta$ LOI parameter has been used to study the variations of the major and some minor element amounts as a function of alteration intensity (Figs. 8 and 9). 211 These variations between altered and reference samples (defined by the lowest  $\Delta LOI$ ) are 212 calculated as follows:  $\Delta SiO_{2sample} = (SiO_{2 sample} \times TiO_{2 reference}/TiO_{2 sample}) - SiO_{2 reference} (wt\%)$ . 213 Because the piece of rocks sampled in the drill cores cannot be at the size of a representative 214 sample, the experimental error is not calculable. This is why we adapted the variation scales to 215 their abundance in the specimens: 0 to 50 for SiO<sub>2</sub> and Al<sub>2</sub>O<sub>3</sub>, -7 to 2 for all the other 216 components. Reported in the same scale,  $\Delta SiO_2$  and  $\Delta Al_2O_3$  show contrasted distribution with 217  $\Delta$ LOI: the first one is systematically negative while the second remains roughly constant. Silica 218 loss and aluminium conservation are commonly reported in water rock interactions due to their 219 large difference of solubility. The weight percent variations are more limited for Fe<sub>2</sub>O<sub>3</sub>, MgO and 220 CaO. If they are not significant for the first and second ones, it seems that the losses of the 221 calcium are more important in the illitic alteration zone. The chemical differences between 222 propylitized and illitized samples are clearly evidenced by the alkaline element behaviour. All 223 exhibit increasing losses of Na<sub>2</sub>O with  $\Delta$ LOI but they more important for the second ones. The 224 K<sub>2</sub>O amount remains roughly constant in the propylitized zone while it significantly increases in 225 the illitic one. The variations for trace elements are expressed in ppm (Fig. 9). The illitized 226 samples are enriched in rubidium and cesium because of their potassium similar chemical 227 affinities for micas. Strontium is depleted in all the samples, the losses being more important in 228

the illitized ones. Cerium and lanthanium amounts are highly variable in the propylitized samples 229 230 while they are systematically depleted in the illitized ones (Figs. 9 and 10). The variations of uranium are erratic but thorium seems to be systematically depleted. In spite of the uncertainties 231 232 due to the limited size of the samples, it is possible to distinguish the geochemical characteristics 233 of the two alteration types. Indeed, the propylitic alteration is almost chemically conservative: the losses of SiO<sub>2</sub> and Na<sub>2</sub>O are limited. This is not the case for illitization which is depleted in all 234 the components except aluminium (conserved) and potassium (enriched). The behaviour of trace 235 elements outlines this difference. 236

#### 237 4.4. Crystallo-chemistry of secondary phases

Chlorite and illite-phengite are present in all the studied samples. It is to be noticed that no dior trioctahedral expandable mixed-layer minerals have been detected in the studied samples (Fig.
11a).

#### 241 4.4.1. Chlorite

XRD patterns of randomly oriented powders show that chlorites are of the IIb polytype. The 242 (001)/(002) peak intensity ratio decreases from depth to top (0.60-0.50 to 0.20) indicating that the 243 relative proportion of Fe and Mg ions changes from the deep propylitic zone to the upper illitic 244 level (Fig. 11b). This is coherent with the micro-chemical analyses of individual chlorite crystals. 245 Most of the half-formula units calculated on a 14 oxygen basis, plot inside the brunsvigite 246 composition field defined by Foster (1960). This is the case for all the chlorites in the propylitic 247 248 zone (Table 2, Table 3, Fig. 12b). The Fe/(Fe+Mg) ratio varies from 0.3-0.5 (propylitic zone) to 0.6 - 0.7 (illitic level). 249

#### 250 4.4.2. Illite-phengite

The XRD patterns of randomly oriented powders indicate that the illite-phengite minerals are 251 252 composed of a mixture of  $2M_1$  and 1M polytypes. They are small sized crystals when replacing the feldspars or infilling the quartz microcracks (5-20µm) but become larger when replacing 253 chlorites (10-50µm). Their half-formula units calculated from microprobe analyses using a 11 254 oxygen basis, are coherent with that of dioctahedral micaceous minerals (Tables 4, 5). Two 255 populations are clearly separated in the  $MR^{3+}$ ,  $2R^{3+}$ ,  $3R^{2+}$  diagram proposed by Velde (1985). The 256 phengitic crystals are found in the altered chlorites as shown by the mixing line while the illitic 257 ones are found in the altered feldspars (Fig. 12a). Both exhibit a quite similar range of 258 Fe/(Fe+Mg) ratio: 0.46-0.61 and to 0.52-0.76 for phengite and illite respectively. However, 259 phengites are always K-richer than illite: 0.84-0.95 and 0.73-0.84 respectively. 260

#### 261 4.4.3. Epidote-allanite

Epidotes are of the pistachite type (Table 6). They form everywhere in the rock, partly replacing primary feldspars or chloritized biotites or precipitating into veinlets. Their Fe/(Fe+Al) ratio varies from 0.30 to 0.42. Allanite crystals, on the contrary, form exclusively in chloritized biotites. They present a rather constant Fe/(Fe+Al) ratio (0.3-0.32). Their main chemical characteristic is the very high Nd, Ce and La contents: 0.16-0.07, 0.38-0.36, 0.25-0.17%respectively.

#### 268 4.4.4. Monazite

Monazite crystals have been observed in two samples: KA6 437.85 (0.85 m from the unconformity) and GR1 631 (1 m from unconformity). The first one is a strongly illitized rock containing Fe-rich chlorites. Euhedral elongated millimeter size crystals of monazite are scattered in the illitic matrix. They seem to be perfectly unaltered and do not exhibit any chemical zoning

in spite of light grey intensity variations observed in the BSE imaging contrast (Table 7). The 273 274 GR1 631 sample is less illitized since only feldspars are altered. Most of the micro-cracks are filled by calcite, and locally by quartz and pyrite. Here, the monazite crystals are anhedral, often 275 grouped in aggregates and intimately associated with sulfide and illite. Compared to the euhedral 276 crystals of the KA6 437.85 sample, monazite crystals of the GR1 631 one exhibit higher U, La 277 and Ce and lower Th, Ca, Nd, Sm contents. The Th/U ratios are very low (Tables 8.9). They are 278 characterized by lower Th content (Th<12000 ppm) compared to that of the unaltered monazites 279 described by Mathieu et al. (2001) in the FA Sandstone (average Th = 40000 ppm for 94 280 analyses). 281

#### 282 4.4.5. Carbonates

Calcite is observed in many samples from depth to the upper levels in the FA sandstone. Far from the unconformity in the propylitic zone, it appears as xenormorphic crystals located in altered plagioclase and in veinlets where it is associated to chlorite. It is observed as euhedral crystals filling the fissures near the unconformity where it is locally associated to anhydrite. In some of the FA sandstone samples, it constitutes the inter-granular microcrystalline cement. Regardless of their habit, the calcite crystals do not show significant chemical composition differences.

#### 290 4.5. Geochronological data

The isotopic dating using laser-ICPMS microanalyses were performed on monazite crystals sampled near the FA sandstone-granitoid unconformity in the KA6 437.85 and GR1631 (Tables 8 and 9). The first one is Th-rich (11614-2909 ppm). Plotted in the <sup>206</sup>Pb/<sup>238</sup>U versus <sup>207</sup>Pb/<sup>235</sup>U diagram (Fig. 13), the U-Pb data of the KA6 437.85 monazite give two groups of discordia ages:

from 2998±25Ma (MSWD = 0.72) to 2922±24 Ma (MSWD = 1.4) and 2621±30Ma (MSWD = 1.5). The GR1 631 monazite being common Pb-rich, the dating is obtained using the  ${}^{207}$ Pb/ ${}^{206}$ Pb versus  ${}^{206}$ Pb/ ${}^{238}$ U. The Tera-Wasserburg diagram (Fig. 14) gives an age of 1870 ± 26Ma (MSWD = 1.04).

299

#### **300 4.6.** C and O isotope compositions of calcite

The calcite isotopic analyses were performed on the propylitized and illitized granitoids far from and closed to the unconformity respectively. The  $\delta^{13}$ C values obtained varies from -5 to -14‰ and the  $\delta^{18}$ O ones from 10.2 and 17.7‰, most of them being close to 10.5 – 11‰ (Tab. 10). The C and O isotope compositions are not inter-correlated, nor with depth, intensity of carbonation, nature of alteration (propylitization, illitization), or element mobility.

The large range of  $\delta^{13}$ C values suggest that carbon comes from two different sources: one 306 having a  $\delta^{13}$ C > -5%, which might be surface-derived continental carbon including atmospheric 307 CO<sub>2</sub>, and another one having a  $\delta^{13}$ C < -14%, which could possibly be a signature of the evolved 308 organic matter similar to that identified in the sediments of the Franceville Basin (Albani et al., 309 2010; Weber and Gauthier-Lafave, 2013). The  $\delta^{18}$ O values centered around 10.5 – 11% point to 310 formation at low temperature hydrothermal conditions (150-300°C), which is consistent with the 311 common mineralogical association of calcite + chlorite, where the fluids reached isotopic 312 equilibrium with the granitic basement. On the other hand, the KA6462 sample has a much 313 higher  $\delta^{18}$ O value, which probably relates to calcite formation at lower temperatures than for the 314 others, regardless of the nature of the fluid involved in calcite crystallization. 315

As a whole, the C and O isotope compositions are consistent with the presence of a diagenetic 316 317 fluid that would have circulated along or close to the unconformity, imposing low temperature hydrothermal conditions to secondary alteration processes as it is the case for the Athabasca basin 318 (Richard et al., 2013). Part of the carbon was taken from the organic matter which is particularly 319 320 abundant in the FB Francevillian sediments. A similar case is given by the Athabasca basin where metamorphic fluids have interacted with graphite bearing metasediments of the basement. 321 These fluids have been able to precipitate calcite with such isotopic signatures, at greenschists 322 ANUS conditions, in equilibrium with chlorite. 323

5. Discussion 324

#### 5.1. Propylitic alteration 325

Sère (1996) interpreted the presence of chlorite, albite, quartz and calcite in veins crosscutting 326 the Chaillu massif as witnessing a metamorphic episode of the greenschist facies type. The 327 328 petrographic observation performed on the eight drill cores show that all the biotite crystals have been chloritized regardless of the parent rock type: granitoids or gneiss. This implies that the 329 330 chloritization took place after deformation. Consequently, it cannot be related to a metamorphic 331 event but rather to a propylitic alteration which preserves the texture of rocks. Indeed, the 332 chlorite-sericite-epidote/calcite assemblage is commonly described in propylitized granitoids (Lowell and Guilbert, 1970; Titley et al., 1986). 333

334 The  $\Delta Fe_2O_3/TiO_2$ ,  $\Delta MgO/TiO_2$ ,  $\Delta CaO/TiO_2$ , and  $\Delta K_2O/TiO_2$  ratios do not vary with  $\Delta LOI$  in the 335 propylitic zone (Fig 10). Some samples plot far from general trend due to the local invasion by secondary minerals such as calcite (high CaO/TiO<sub>2</sub> ratio-GR23 631) in fracture and plagioclase. 336 337 The  $\Delta SiO_2/TiO_2$  and  $\Delta Na_2O/TiO_2$  ratios decrease slightly while  $\Delta Al_2O_3/TiO_2$  remains roughly

constant. This means that the propylitized rocks do not have experienced significant exchanges of 338 339 chemical component with the external reservoir. It is a nearly isochemical set of mineral reactions in which the potassium released by biotite during chloritization is consumed by the crystallization 340 of sericite. This is typical of propylitic alteration in which the water/rock ratio is limited (Berger 341 342 and Velde, 1992). The temperature conditions can be roughly estimated using mineral indicators. The allanite is known to be found at rather high temperature: 400 °C (Berger and Velde, 1992; 343 Smith and Barreiro, 1990; Wing et al., 2003) while the chlorite-epidote-paragenesis is formed in 344 the range 350 to 200 °C (Norman et al., 1991). These conditions are compatible with that 345 prevailing during the cooling stage following the intrusion of plutonic rocks. 346

#### 347 **5.2 Illitic alteration**

Petrographic observations suggest that the propylitic alteration is overprinted by an illitic one 348 in the upper part of the basement within 4-5 m below the unconformity. It is typically 349 characterized by a massive replacement of all the primary and secondary minerals except the 350 sericite zones in feldspars which seem to have been preserved. This could explain the co-351 352 existence of the  $2M_1$  and 1M polytypes of the white micas fraction. The degree of replacement decreases with depth down to 7 m where illite is observed only in veins. Epidote and allanite 353 have totally disappeared while chlorite seems to remains up to the unconformity. However, its 354 composition changes becoming iron-richer composition. This indicates that it has recrystallized 355 356 in a more stable version considering the local chemical condition imposed by the fluid responsible of the illitic alteration. As shown by the strong decrease of the  $\Delta CaO$  and  $\Delta Na_2O$ 357 358 ratios on one hand and the increase of the  $\Delta K_2O$  ones on the other hand, it appears that the illitic alteration induces more important chemical changes of the rocks that the propylitic one did (Fig. 359 360 8). It is obvious that it operates in an open system in which the fluids have flowed. The mineral

similarities between illite in the most altered basement samples and the upper sandstones militate for extensive chemical exchanges driven by the diagenetic fluids along the unconformity. The very low thorium content of the small euhedral monazite crystals indicates that they have crystallized at low temperature (Overstreet, 1960) and are probably synchronous with the illitic alteration.

#### **5.3.** Could a paleosol have been present along the unconformity?

Based on the study of fifty pre-Devonian rock units that have been described as paleosols, Rye and Holland (1998) proposed that five criteria must be respected to evidence the weathering effects: *(i)* the paleosol is developed on a homogeneous parent rock and has been preserved in place; *(ii)* textural; *(iii)* mineralogical; *(iiii)* chemical; *(iiiii)* soft sediments features. A geological formation that meets all these criteria is classified as definite paleosol; when all criteria are met but one it is likely a paleosol; all but two is a possible paleosol; not more than three criteria is unlikely a paleosol.

Considering the eight drill cores studied here, no petrographic features typical of a weathering 374 process have been observed, even in the reddish samples from the GR31 or the GR20 drill cores. 375 376 None of them contain any evidence of typical weathering processes such as primary phase recrystallization inside microsytems, clay translocation deposits (cutans) or increasing porosity 377 378 toward the surface (Velde and Meunier, 2008). On the contrary, the rocks, irrespective of their petrographic properties, are all invaded by a single clay mineral phase: illite. If any physical 379 380 traces of a weathering event have been preserved after the sediment deposition, they have been totally erased by the diagenetic alteration processes. 381

As it is the case for any water/rock interactions induced by pure water or diluted solution 382 383 invasion, the most soluble components are leached out from the altered rock (Ca, Na, Mg) while the less soluble (Al, Ti) ones are concentrated and K is enriched (Fig. 8). These chemical 384 characteristics do not meet the criteria proposed by Rye and Holland (1998). Consequently, the 385 386 probability for a paleosol to have been formed on the unconformity surface before being buried under the sediments of the Franceville basin is very weak. The superimposition of an illitic 387 alteration upon a previous propylitic one is the most probable scenario here. It is supported by the 388 distinct chemical trends evidenced by plotting the bulk rock compositions in the Al<sub>2</sub>O<sub>3</sub>-CaO, 389 Na<sub>2</sub>O, K<sub>2</sub>O-FeO, MgO system (Fig. 15). Two trends are clearly separated: (i) a shift toward the 390 epidote-chlorite assemblage typical of the propylitic alteration or magmatic differentiation; (ii) a 391 shift toward the illite pole. The two different chemical pathways are coherent with the 392 progressive upward variation of Rb, Cs and Sr amounts which are increasing and decreasing 393 respectively (Fig. 9). The same contrasted chemical behavior is also observed for the REE 394 distribution: the amounts of the light ones (LREE) do not vary during the propylitic alteration 395 while they are leached out during the illitic event (Fig. 9 and 10). This is also coherent with the 396 disappearance of the allanite crystals which could be related to the percolation of oxidizing 397 diagenetic fluids invading the basement from the overlying sedimentary basin through the 398 unconformity (Cuney and Mathieu, 2000). 399

400 The negative  $\delta^{13}$ C values of calcite crystals in veins indicate that the diagenetic fluids involved 401 in the illitization process were mixed with organic compounds deriving from oil maturation in the 402 FB sediments (Gauthier-Lafaye and Weber, 1989; Weber and Gauthier-Lafaye, 2013).

403 **5.4. Timing of alteration events.** 

The isotopic U-Pb data gathers into three major dating groups. The oldest one, from  $2998 \pm 25$ 404 405 Ma to  $2922 \pm 24$  Ma is consistent with the age of plutonic intrusions forming the basement. The 406  $2621 \pm 30$  Ma intermediate group might correspond to the hydrothermal propylitic event related to the intrusion of the neoarchean plutonic bodies. It is to be noticed that the dating obtained here 407 408 from monazite crystals is coherent with that measured in altered biotites using the K-Ar method:  $2696 \pm 60$  Ma (Bonhomme et al., 1978). The youngest group, i.e.  $1870 \pm 26$  Ma, has been 409 measured on newly formed monazites from the illitized zone. Their extremely low Th contents 410 reflect a crystallization process under low temperature conditions. It is consistent with the 411 diagenetic event of the Franceville basin (Bonhomme et al., 1982). 412

#### 413 **6.** Conclusion

The mineralogical and geochemical studies of the basement of the Franceville Basin along a 414 nearly E-W profile show that the basement is composed of different plutonic rocks. Dating 415 obtained from 8 drill holes give Archean ages (3.0 - 2.9 Ga) similar to the ones obtained on rocks 416 outcropping in the Chaillu Massif. No metasedimentary formations have been intersected. Most 417 of these rocks present evidences of a propylitic alteration which postdate any ductile deformation 418 and which may be associated to the intrusion of some Neoarchean plutonic rocks at about 2.6 to 419 2.7 Ga (Thiéblemont et al., 2009). The mineral reactions produced a Mg chlorite + epidote + 420 421 allanite + titanite paragenesis typical of nearly close systems: no significant mobility of the major elements suggesting weak Water/Rocks ratios. Then, a massive illitic alteration associated with 422 hematization, has transformed the previously propylitized rocks over a thickness of 2 to 5 m 423 below the unconformity in the Franceville basin. It has been triggered by the percolation of 424 425 oxidizing diagenetic fluids from the FA conglomeratic-sandstone formation at the base of the 426 sedimentary basin through the unconformity. Similar fluid invasions have been described in other

basins: Southern Norway (Munz et al., 1995; Oliver et al., 2006), Spain (Bouch et al., 2006), 427 France, Northwestern Massif Central (Munoz et al., 1999), Athabasca, Canada (Richard et al., 428 2013) and Kombolgie, Australia (Derome et al., 2003). Most of the major and light Rare Earth 429 elements have been heavily depleted. This is particularly the case for Ca, Na and Mg while Al 430 amounts are roughly preserved. No paleosol chemical or petrographical traces have been detected 431 here. The calcite  $\delta^{13}$ C values confirm the diagenetic origin for the fluids which have percolated 432 into the basement. They were also responsible of the uranium transfer and deposition in the 433 vicinity of the organic-rich sediments of the FB formation which gave the natural nuclear reactors 434 (Gauthier-Lafaye and Weber, 2003; Mathieu et al., 2000). 435

436

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- 610

611 **Figure captions** 

- 612
- **Fig. 1**: Simplified geological map of Franceville basin and location of the studied region

614 (modified after Bros et al, 1993).

615

Fig. 2: (a) Location of the studied drill cores (KA13, GR1, GR5, GR20, GR23, GR31, GR 43 and
KA6) close to Kaya-Kaya fault in the Franceville basin; (b) Simplified geological logs of the
studied drill cores.

- 619
- **Fig. 3**: Bulk rock chemical compositions in the Debon and Le Fort diagram (1983). In brackets (L.O.I.), Q = normative quartz, P = normative feldspars. (a) GR20 and GR31 drill cores, (b) GR23, KA6 drill cores. Dot arrow = magmatic differentiation trend. The white triangles represent the average compositions of the diorite quartzite (dq), tonalite (to), granodiorites (gd), and adamellite (ad)
- 625

Fig. 4: Chondrite-normalized rare earth element patterns of the least altered rocks.
Adamellite(KA6 472), Granodiorite (KA6 445, GR31 873 and GR31 866), Tonalite (KA6 467,
KA6 454, KA6 449, KA6 444, KA6 438, GR23 640), Diorite (KA6 462).

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**Fig. 5**: Photomicrographs of the granitoids affected by propylitic alteration: (**a**) biotite is completely replaced by the chlorite associated to rod shaped allanite and titanite, backscattered electron SEM image; (b) sericitization of the feldspar, chloritization of biotite and chlorite calcite sequence of fissure infilling. chloritized biotite = chl biot, chlorite = chl, allanite = All, titanite = tin, feldspar = Felds, calcite = Cal, QZ = quartz.

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Fig. 6: Photomicrographs of illitization features in granitoids; (a) feldspars are completely
replaced by illite while quartz (Qz) are cracked and enveloped by illitic matrix; (b) The
chloritized biotite is partially to totally replaced by the Fe-oxide + phengite (Phg) assemblage; (c)
fracture filled by calcite (cal); (d) Cementation in the FA sandstone by anhydrite.

640

641 **Fig. 7**: Evolution of  $\Delta$  LOI with depth of tonalite.

642

**Fig. 8**: Plot of  $\Delta$ SiO<sub>2</sub>,  $\Delta$ Al<sub>2</sub>O<sub>3</sub>,  $\Delta$ CaO,  $\Delta$ Na<sub>2</sub>O,  $\Delta$ Fe<sub>2</sub>O<sub>3</sub>,  $\Delta$ MgO,  $\Delta$ K<sub>2</sub>O versus  $\Delta$ LOI of

644 propylitized (black diamond) and illitized (open circle) tonalite.

**Fig. 9**: Plot of ΔRb, ΔSr, ΔLa, ΔCe, ΔCs, ΔTh, ΔU versus ΔLOI of propylitized (black diamond) and illitized (open circle) tonalites.

647

Fig. 10: Evolution of the chondrite-normalized rare earth element patterns of the granitoids, from
the least altered to the most altered (illitized) granitoids. Open square = illitized granitoids, black
losange = superimposed illitic on propylitic alteration, open circle = least altered and propylitized
granitoids.

652

**Fig. 11**: (**a**) Evolution of the XRD patterns of the < 2  $\mu$ m fractions from KA6 467 m (granitoids) to KA6 436 m (sandstone); (**b**) Evolution of the XRD patterns of chlorite from KA6 467 m tonalite), to KA6 438 drill core < 2 $\mu$ m fraction in oriented glycol saturation preparations. Chl = chlorite, ill = illite.

657	
658	Fig. 12: (a) Representation of the four clay mineral domains in the MR <sup>3+</sup> , 2R <sup>3+</sup> , 3R <sup>2+</sup> triangle
659	(Velde, 1985): illite, phengite (Ph), chlorite upper zone (chlup) and chlorite lower zone (chl <sub>low</sub> ),
660	phengite + chlorite mixture (Ph + $chl_{up}$ ) domain. (b) Evolution of chemical compositions of the
661	chlorites analyzed from the depth (lower zone) to the unconformity (upper zone) in the Fe/(Fe +
662	Mg) versus Si diagram. Open diamond = chlorite of lower zone, open circles = chlorite of upper
663	zone.
664	
665	Fig. 13: Representation of Pb and U isotopic data in the <sup>206</sup> Pb/ <sup>238</sup> U versus <sup>207</sup> Pb/ <sup>235</sup> U diagram.
666	MSWD = Mean Square Weighted Deviation. Monazites of KA6 437.85 sample.
667	
668	Fig. 14: Representation of Pb and U isotopic data <sup>207</sup> Pb/ <sup>206</sup> Pb versus <sup>206</sup> Pb/ <sup>238</sup> U diagram. MSWD
669	= Mean Square Weighted Deviation. Monazites of GR1 631 sample.
670	
671	Fig. 15: (a) Representation of the alteration effects in the $Al_2O_3 - CaO^* + Na_2O + K_2O - FeO + CaO^* + C$
672	MgO diagrams (Rainbird et al., 1990). CaO* = CaO of silicates. L.alt = least altered rocks, Pro +
673	dif = propylitized rocks + magmatic differentiation, Ill. = analyzed illites, phg = analyzed
674	phengites. Illite = theorical illite, Ka = kaolinite, Gi = Gibbsite. Black arrows indicate the least
675	altered rock to-propylitized rock and the propylitized rock-to- illitized rock. (1) 2.6 Ga propylitic
676	event, (2) 1.87 Ga late diagenetic (illitic) event.
677	
678	Table captions
679	
680	Table 1. Whole-rock geochemical data for the Archean granitoids at Kiéné area.
681	
682	Table 2: Representative microprobe analyses of chlorites from the propylitic zone. The structural
683	formulae are calculated on the 14 oxygen basis.
684	
685	<b>Table 3</b> : Representative microprobe analyses of chlorites from the illitic zone. The structural
686	796 formulae are calculated on the 14 oxygen basis.
687	

688	Table 4: Representative microprobe analyses of phengites from 797 the illitic zone. The
689	structural formulae are calculated on the 11 oxygen basis.
690	
691	<b>Table 5</b> : Representative microprobe analyses of illites from the illitic zone. The structural
692	800 formulae are calculated on the 11 oxygen basis.
693	
694	<b>Table 6</b> : Representative microprobe analyses of epidotes and allanites from the propylitic zone.
695	The structural formulae are calculated on the 12.5 oxygen basis.
696	
697	Table 7: Representative microprobe analyses of monazites.
698	
699	Table 8: Representative Laser-ICPMS analyses and age calculations of monazites (sample
700	805 KA6 437.85).
701	
702	Table 9: Representative Laser-ICPMS analyses and age calculations of monazites (sample
703	807 GR1 631).
704	
705	Table 10: $\delta^{13}$ C values (% PDB) and $\delta^{18}$ O (% SMOW) measured on six samples affected by
706	carbonated alteration.
707	
708	
	· · · · · · · · · · · · · · · · · · ·



Figure 1



Figure 2



Figure 3



Figure 4



# Figure 5



# Figure 6



Figure 7



Figure 8

![](_page_39_Figure_1.jpeg)

Figure 9

![](_page_40_Figure_1.jpeg)

## Figure 10

![](_page_41_Figure_1.jpeg)

Figure 11

![](_page_42_Figure_1.jpeg)

![](_page_43_Figure_1.jpeg)

## Figure 13

![](_page_44_Figure_1.jpeg)

Figure 14

![](_page_45_Figure_1.jpeg)

Figure 15

Table 1

	Granitoid				Sandstone			Granitoid				Sandstone	
-	GR23 640,50	GR23 631,50	GR23 630,90	GR23 629,90	GR23629,25	GR23 629,18	GR20 851	GR20845	GR20 833	GR20 825	GR20821	GR20819	GR20818
SiO <sub>2</sub> (Wt%)	64,74	63,91	61,72	63,42	57,00	70,17	69,05	68,08	56,74	60,99	62,10	55,55	75,20
TiO <sub>2</sub>	0,45	0,50	0,39	0,44	0,56	0,09	0,30	0,19	0,32	0,52	0,38	0,70	0,07
$Al_2O_3$	16,15	15,07	15,81	15,58	22,83	5,17	15,35	16,49	18,45	17,66	18,62	23,36	5,04
FeOt	4,23	4,01	3,93	4,07	4,08	0,87	2,65	1,86	8,16	4,71	4,33	6,86	0,62
MnO	0,05	0,06	0,07	0,02	0,02	0,00	0,03	0,02	0,06	0,07	0,04	0,01	0,00
MgO	3,15	2,05	1,88	1,22	1,46	0,31	1,79	0,89	1,61	2,06	2,01	0,73	0,26
CaO	2,24	4,19	5,07	2,67	0,58	8,17	2,03	3,07	6,68	5,32	1,12	0,69	6,50
Na <sub>2</sub> O	5,13	4,66	5,49	2,42	0,35	0,07	5,21	5,84	4,66	5,46	2,30	0,14	0,03
K <sub>2</sub> O	1,89	1,80	1,42	4,50	8,34	2,32	1,23	1,43	0,59	0,71	5,01	7,38	1,68
$P_2O_3$	0,23	0,26	0,23	0,27	0,03	0,00	0,10	0,08	0,07	0,14	0,08	0,19	0,04
LOI	1,95	3,53	4,17	4,52	4,03	12,20	1,65	1,55	2,09	2,03	3,26	3,94	9,10
Total	100,20	100,03	100,18	99,12	99,25	99,37	99,38	99,50	99,41	99,67	99,24	99,55	98,55
Ba(ppm)	1060,00	1147,00	476,40	4229,00	1700,00	830,40	401,20	622,10	243,70	388,30	2552,00	831,90	469,50
Rb	40,77	70,11	61,46	261,50	433,60	107,80	30,21	32,20	15,80	17,39	247,00	259,20	62,30
Sn	1,12	1,18	1,07	0,97	1,91	0,32	0,59	0,00	0,91	1,16	0,87	1,93	0,31
Sr	802,90	695,90	533,40	775,50	115,50	780,00	490,10	738,60	1293,00	573,20	204,50	15,23	385,50
Ga	21,62	21,71	20,30	20,46	33,50	6,43	17,50	17,85	26,20	22,31	23,63	29,91	6,09
Y	9,72	11,63	11,09	12,01	25,19	7,12	3,76	3,66	9,05	14,35	7,46	13,02	3,71
Zr	141,80	179,50	137,30	155,50	221,00	337,00	59,26	100,80	368,40	158,80	32,17	286,50	124,20
Nb	4,07	3,61	2,83	3,54	7,49	2,29	2,10	1,32	1,04	3,62	2,73	7,48	1,83
Cs	0,50	1,89	1,43	4,49	16,31	1,52	0,42	0,33	0,33	0,27	4,32	17,79	3,04
Cu	0,00	47,83	93,11	20,02	8,18	10,75	7,91	23,41	5,25	15,86	39,96	4,06	22,26
Zn	6/,14	66,51	99,82	93,88	81,95	19,63	43,66	33,26	/6,9/	66,83	110,70	39,94	15,07
HI T-	3,59	4,34	3,35	3,82	5,39	1,18	1,05	2,87	8,81	3,82	1,00	6,99	3,17
1a Dh	0,26	0,23	0,20	0,44	0,45	0,25	0,14	0,08	0,11	0,35	0,20	0,42	0,17
F U Th	7,20	42,05	2 20	25,25	19,27	5,44 11.96	0,02	10,04	0,72 7,52	10,20	5,24	0,23 20.04	1,95
III II	0.50	1,05	2,20	41,43	12,43	3 50	2,85	0,02	2,50	0.34	4,50	20,04	1.22
V	64.36	74.24	70.28	4,55 62.46	256 10	146.40	40.54	14 51	80.11	73 27	51.95	84.92	10.02
v La	33.46	65.07	58.12	39.61	114.90	14.42	11.47	12.42	22 37	11.01	17.16	20.31	3.40
Ce	59.83	119.40	105 20	75.67	180.30	24 56	22 37	25 32	41 44	25.44	29.71	41 50	5.93
Pr	676	12.87	11 33	9.08	19.94	3.15	2 71	3.01	4 23	3 42	3 42	4 77	0.75
Nd	24.99	46 31	41.11	32.99	74.04	12.13	10.30	11.32	13.56	14 44	2.73	17.84	2.85
Sm	4.28	7.29	6.41	5.78	12.69	2.40	1.72	1.92	2.00	3.18	2.85	2.94	0.60
Eu	1.36	1.68	1.35	1.87	2.98	0.66	0.56	0.53	0.61	0.73	0.90	0.64	0.21
Gd	3,02	4,46	3,94	4,15	9,36	1,84	1,23	1,11	1,61	2,77	2,15	2,62	0,70
Tb	0,37	0,52	0,48	0,54	1,21	0,24	0,16	0,14	0,22	0,43	0,30	0,37	0,10
Dy	1,80	2,44	2,25	2,65	5,90	1,19	0,75	0,66	1,24	2,57	1,64	2,14	0,63
Ho	0,31	0,39	0,37	0,42	0,89	0,22	0,13	0,12	0,26	0,49	0,29	0,45	0,12
Er	0,80	1,02	0,99	1,05	2,07	0,67	0,35	0,32	0,78	1,41	0,77	1,41	0,34
Tm	0,11	0,14	0,14	0,14	0,26	0,10	0,05	0,04	0,12	0,21	0,11	0,22	0,05
Yb	0,72	0,93	0,89	0,87	1,53	0,78	0,16	0,14	0,22	1,36	0,72	0,37	0,34
Lu	0,11	0,14	0,14	0,13	0,19	0,14	0,06	0,05	0,15	0,21	0,10	0,26	0,06
			,0										

#### Table 1 (Continued)

			Granitoid			Sandstone					Granito	d					Sandstone
	GR31873	GR31866	GR31863	GR31859	GR31858	GR31857	KA6472	KA6467,90	KA6462	KA6454	KA6449	KA6445,60	KA6444	KA6438,7	KA6437,85	KA6437	KA6432
SiO <sub>2</sub> (Wt %)	73,73	74,47	62,19	73,33	66,75	86,53	73,68	69,54	52,11	65,63	65,07	74,17	65,20	71,01	60,91	56,67	92,05
TiO <sub>2</sub>	0,14	0,23	0,62	0,13	0,21	0,09	0,04	0,33	0,99	0,45	0,40	0,04	0,47	0,42	0,61	0,87	0,05
$Al_2O_3$	14,68	12,78	17,78	14,37	20,33	6,78	13,92	14,18	17,26	16,52	16,27	13,81	15,24	15,27	22,43	22,71	3,92
FeOt	1,76	2,80	5,46	2,39	1,44	0,70	0,30	3,32	9,41	4,03	4,86	0,47	4,58	2,40	2,76	6,20	0,20
MnO	0,02	0,02	0,10	0,01	0,03	0,01	0,01	0,03	0,09	0,03	0,04	0,01	0,05	0,02	0,01	0,03	0,00
MgO	0,38	0,43	2,38	0,74	0,59	0,27	0,17	1,45	3,50	1,47	1,59	0,16	1,82	0,66	0,91	1,67	0,10
CaO	1,14	1,21	0,75	0,32	0,16	0,44	0,89	1,27	4,26	1,59	1,61	1,47	1,53	1,75	0,42	0,06	0,04
Na <sub>2</sub> O	4,85	4,27	3,61	1,93	0,09	0,05	4,03	5,81	4,43	5,59	5,53	4,61	5,18	5,00	0,07	0,09	0,12
$K_2O$	3,02	2,39	3,82	4,18	6,11	2,19	4,92	1,48	2,32	2,25	1,93	3,31	1,90	2,29	7,40	7,25	1,97
$P_2O_3$	0,03	0,08	0,19	0,03	0,03	0,06	0,00	0,19	0,82	0,23	0,23	0,00	0,21	0,22	0,28	0,03	0,00
LOI	0,76	0,95	2,82	1,72	3,52	1,77	0,99	1,42	4,26	1,78	2,12	1,36	2,64	1,67	3,90	3,89	0,51
Total	100,51	99,61	99,72	99,14	99,24	98,88	98,93	99,02	99,44	99,57	99,63	99,40	98,81	100,71	99,69	99,47	98,95
<b>B</b> a( <b>ppm</b> )	1014.00	740 50	577 10	874 60	528 40	1800.00	3861.00	076 40	1355.00	2100.00	1551.00	5212.00	1100.00	000.40	2007.00	2227.00	543 70
ва(ррш) ВЪ	61 78	60.31	194.40	182 50	168.60	74 69	95.06	25.94	70.29	62.86	54 00	59.10	63 20	71.04	2007,00	2237,00	50.84
Sn	0.62	0.72	1 54	0.58	0.75	0.00	0.00	1.05	2.07	1 20	1 33	0.00	3 23	0.89	240,80	1 80	0.00
Sr	222.40	202.30	121.80	59.10	15.80	131.80	553.80	658 20	643 50	650 70	587.60	634.60	370.10	586 30	57.23	52.65	32.24
Ga	17.28	17.18	24.91	20.08	26.15	8 77	13 20	16 31	24.81	22 77	23-01	14 97	23.95	18 14	31.57	28.22	4 31
v	4 62	13.04	8 68	12 32	11.28	3.97	0.80	9.95	35 31	11.16	11 25	2 02	9.67	8 55	11.46	15 69	2 11
7 7r	57.04	165.80	110.60	97 47	167 30	47.66	16.28	120 30	234 30	154.00	146 90	50.36	160 50	150 50	218 10	262.70	59.63
Nb	2.91	5 71	3.92	2.97	5 50	1 34	0.21	3 61	8 33	5.07	5.08	0.23	5 39	4 86	6.21	10.41	1 31
Cs	0.55	0.71	3 40	2.72	8.01	4 47	0.55	0.15	1.85	0.79	0.91	0.39	1.25	2.14	15 50	7.05	1 10
Cu	6 54	11 37	34.02	4 29	0.00	12.30	8 30	17.85	65 30	42.13	18.92	123.80	41 43	0.00	0.00	0.00	5 55
Zn	30.30	41.49	268.60	74.10	59.29	29.28	0.00	46.25	130.00	50.28	62.36	0.00	58.56	32.64	39.11	73.37	0.00
Hf	1.61	4.71	3.05	3.43	5,39	1.33	0.38	2,96	5.47	3.79	3.65	1.60	3.98	3.63	5.42	6.92	1.63
Та	0,15	0,35	0,27	0,08	0,24	0,12	0,01	0,21	0,47	0,33	0,34	0,02	0,34	0,43	0,67	1,00	0,13
Pb	7,25	6,66	5,70	10,27	3,43	2,07	8,99	4,04	29,46	45,59	16,15	10,68	45,39	88,14	5,57	10,07	4,50
Th	1,38	8,79	2,09	4,37	9,97	4,13	0,77	15,08	17,55	10,46	6,38	0,67	12,69	8,79	7,52	24,00	3,45
U	0,51	0,89	1,49	1,99	2,78	1,35	0,23	2,78	4,98	3,97	2,78	0,61	3,97	4,83	2,83	12,94	0,60
V	10,49	18,94	79,74	67,65	14,88	14,27	4,92	86,66	199,90	116,20	128,40	4,08	88,59	37,92	54,50	75,21	2,50
La	6,82	11,55	14,41	25,61	4,92	17,36	1,45	34,16	103,70	67,71	49,74	10,40	45,73	25,89	55,00	11,44	11,30
Ce	12,31	23,10	28,22	46,72	10,56	28,32	2,25	63,12	213,90	124,50	103,90	17,91	88,34	50,81	101,30	22,80	16,67
Pr	1,26	2,72	3,44	5,22	1,17	3,49	0,23	7,55	24,13	14,27	11,01	1,85	9,58	5,97	10,45	2,88	1,99
Nd	2,91	5,71	3,92	2,97	4,60	11,81	0,85	27,66	96,64	51,54	40,18	6,34	5,39	22,82	45,21	12,33	6,61
Sm	0,94	2,21	2,96	3,66	1,33	2,07	0,17	4,79	16,82	7,69	6,53	0,85	6,65	3,99	8,11	3,55	0,90
Eu	0,61	0,76	0,89	0,95	0,45	0,52	0,21	3,15	11,24	4,50	3,93	0,56	1,21	2,83	1,57	1,33	0,21
Gd	0,91	2,30	2,69	2,92	1,56	1,65	0,15	0,41	1,42	0,49	0,47	0,06	4,01	0,34	5,05	3,41	0,55
Tb	0,15	0,38	0,35	0,41	0,25	0,20	0,02	2,10	7,20	2,27	2,32	0,32	0,46	1,70	0,58	0,50	0,07
Dy	0,83	2,24	1,78	2,23	1,62	0,86	0,11	2,10	7,20	2,27	2,32	0,32	2,14	1,70	2,67	2,73	0,38
Ho	0,15	0,43	0,27	0,40	0,35	0,13	0,02	0,35	1,25	0,37	0,39	0,06	0,34	0,29	0,43	0,48	0,06
Er	0,40	1,14	0,62	1,04	0,96	0,30	0,07	0,96	3,40	0,97	1,08	0,18	0,91	0,81	1,15	1,27	0,19
Tm	0,06	0,17	0,08	0,15	0,14	0,04	0,01	0,13	0,48	0,14	0,15	0,03	0,13	0,11	0,17	0,19	0,03
Yb	0,36	1,16	0,47	0,99	1,01	0,24	0,07	0,87	3,06	0,94	1,01	0,20	0,83	0,78	1,14	1,23	0,17
Lu	0,06	0,19	0,07	0,16	0,17	0,04	0,01	0,13	0,48	0,15	0,15	0,03	0,12	0,13	0,18	0,19	0,03
			C	6													

$SIO_2$ (wt %)	27.42	27.22	27.10	28.00	20 (0	21.09	29.70	27.55	20.0
TIO	27,42	27,55	27,19	28,99	28,09	51,08	28,79	27,55	29,0
$10_2$	16.35	0,04 16.10	16.31	10.32	18 53	18.46	10.63	10.00	10,1
Al <sub>2</sub> O <sub>3</sub>	10,55	14.02	13.80	19,52	21.88	10,40	22.05	25 72	25 (
MnO	0.20	0.16	0.20	0.65	0.40	0.12	0.74	0.35	23,2
MgO	19.94	20.13	19.78	17.21	17 27	18 71	17 33	15 54	14
CaO	0.09	0.12	0.02	0.03	0.20	0.07	0.05	0.06	0.0
NacO	0,09	0,12	0,02	0,03	0,29	0.05	0,05	0.02	
K <sub>2</sub> O	0,05	0,04	0.02	0,03	0.04	0,05	0.04	0.04	0,0
Total	82,66	0,10 79,01	0,00 77,50	89,07	87,92	88,40	89,74	88,39	89,2
Si	2,97	3,03	3,05	2,95	2,95	3,12	2,91	2,88	2,9
Al <sup>IV</sup>	1,03	0,97	0,95	1,05	1,05	0,88	1,09	1,12	1,0
Al <sup>VI</sup>	1,05	1,14	1,21	1,27	1,20	1,30	1,26	1,23	1,3
Ti	0,00	0,00	0,00	0,00	0,06	0,01	0,02	0,00	0,0
Fe <sup>2+</sup>	1,67	1,38	1,30	1,94	1,88	1,65	1,94	2,25	2,2
Mg	3,22	3,32	3,31	2,61	2,65	2,80	2,61	2,42	2,2
Mn	0,02	0,01	0,02	0,06	0,04	0,01	0,06	0,03	0,0
Ca	0,01	0,01	0,00	0,00	0,03	0,01	0,01	0,01	0,0
Na	0,02	0,01	0,00	0,01	0,01	0,01	0,00	0,00	0,0
K	0,01	0,01	0,01	0,01	0,01	0,01	0,01	0,01	0,0
$\sum \text{oct}$	5,94	5,84	5,83	5,82	5,73	5,74	5,81	5,90	5,7
Fe/(Fe+Mg)	0,34	0,29	0,28	0,43	0,42	0,37	0,43	0,48	0,5
Fe/(Fe+Mg)	0,34	0,29	0,28	0,43	0,42	0,37	0,43	0,48	0,

#### Table 3

		Ch	lorite in the il	litic zone			
SiO <sub>2</sub> (Wt %)	27,65	26,27	26,40	26,18	25,47	26,38	27,97
TiO <sub>2</sub>	2,05	2,10	1,83	1,35	0,03	0,04	0,19
$Al_2O_3$	15,62	15,58	15,85	15,51	17,79	18,77	21,39
FeOt	31,22	31,48	29,90	32,43	31,55	29,70	34,02
MnO	0,34	0,21	0,46	0,40	0,42	0,37	0,16
MgO	10,27	10,70	10,44	10,18	10,57	10,79	5,82
CaO	0,08	0,07	0,09	0,13	0,06	0,03	0,02
Na <sub>2</sub> O	0,05	0,02	0,07	0,04	0,06	0,01	0,00
K <sub>2</sub> O	0,20	0,13	0,15	0,44	0,33	0,64	0,91
Total	87,48	86,55	85,18	86,65	86,29	86,73	90,47
Si	3.02	2.92	2.96	2.93	2.85	2.89	2.87
Al <sup>IV</sup>	0.98	1.08	1.04	1.07	1.15	1.11	1.13
Al <sup>VI</sup>	1,04	0,96	1,05	0,98	1,19	1,32	1,47
Ti	0,17	0,18	0,15	0,11	0,00	0,00	0,02
Fe <sup>2+</sup>	2,86	2,93	2,80	3,04	2,95	2,72	3,22
Mg	1,67	1,77	1,74	1,70	1,76	1,76	1,03
Mn	0,03	0,02	0,04	0,04	0,04	0,03	0,03
Ca	0,01	0,01	0,01	0,01	0,01	0,00	0,00
Na	0,01	0,00	0,02	0,01	0,01	0,00	0,06
K	0,03	0,02	0,02	0,06	0,05	0,09	0,06
∑oct	5,57	5,66	5,60	5,72	5,90	5,80	5,72
Fe/(Fe+Mg)	0,63	0,62	0,62	0,64	0,63	0,61	0,76

<u>0,63</u><u>0,62</u>

$SiO_2$ (wt %) TiO <sub>2</sub>	10 5 1	4616	16.05	17.05	50.70	46.00	47.04	47.00	47.4
11O <sub>2</sub>	48,54	46,16	46,05	47,05	50,79	46,02	47,94	47,29	4/,4
	0,09	0,09	0,12	0,15	0,08	0,10	0,02	0,10	0,1
$Al_2O_3$	26,03	26,84	26,90	25,19	27,04	26,45	28,16	26,72	27,8
FeOt	5,09	5,72	5,43	5,14	4,38	5,99	4,45	5,03	4,6
MnO	0,03	0,08	0,03	0,03	0,12	0,12	0,01	0,05	0,0
MgO	2,07	2,09	1,94	1,79	2,62	1,89	1,97	2,01	1,9
CaO	0,05	0,02	0,09	0,12	0,15	0,04	0,04	0,00	0,0
Na <sub>2</sub> O	0,06	0,03	0,08	0,25	0,07	0,12	0,09	0,03	0,0
K <sub>2</sub> O	10,15	10,14	9,90	9,46	9,91	10,24	10,60	10,57	10,7
Total	92,10	91,17	90,54	89,16	95,15	90,97	93,29	91,79	92,9
Si	3,36	3,25	3,26	3,36	3,38	3,26	3,28	3,30	3,2
Al <sup>IV</sup>	0,64	0,75	0,74	0,64	0,62	0,74	0,72	0,70	0,7
Al <sup>VI</sup>	1,49	1,48	1,50	1,49	1,51	1,47	1,55	1,50	1,5
Ti	0,00	0,00	0,01	0,01	0,00	0,01	0,00	0,01	0,0
Fe <sup>2+</sup>	0,27	0,30	0,29	0,28	0,22	0,32	0,23	0,26	0,2
Mg	0,21	0,22	0,20	0,19	0,26	0,20	0,20	0,21	0,2
Mn	0,00	0,00	0,00	0,00	0,01	0,01	0,00	0,00	0,0
Ca	0.00	0.00	0,01	0,01	0.01	0.00	0.00	0.00	0.0
Na	0,01	0,00	0,01	0,03	0.01	0,02	0,01	0.00	0.0
К	0.90	0.91	0.89	0.86	0.84	0.92	0.93	0.94	0.9
Inter Charg	0.91	0.92	0.91	0.91	0.86	0.94	0.94	0.95	0.9
Σoct	1.97	2.00	2.00	1.95	1.99	1.99	1.99	1.98	1.9
Ee/(Fe+Mg)	0.55	0.58	0.59	0.59	0.46	0.61	0.53	0.56	0.5

#### Table 5

				Illite				
SiO <sub>2</sub> (Wt %)	53,18	48,09	52,96	48,03	47,47	52,21	53,30	48,89
TiO <sub>2</sub>	0,29	0,08	0,09	0,05	0,01	0,09	0,11	0,14
$Al_2O_3$	35,37	34,46	36,45	34,06	32,50	36,27	36,23	32,59
Fe2O3	2,73	1,07	1,81	1,29	1,91	1,83	1,81	1,09
MnO	0,04	0,01	0,00	0,00	0,00	0,04	0,04	0,00
MgO	1,02	0,66	0,69	0,83	0,52	0,55	0,83	1,49
CaO	0,03	0,05	0,04	0,06	0,18	0,04	0,03	0,08
Na <sub>2</sub> O	0,08	0,14	0,08	0,12	0,07	0,08	0,07	0,11
K <sub>2</sub> O	9,51	9,54	9,37	9,81	8,74	9,45	9,40	9,94
Total	102,25	94,08	101,48	94,25	91,40	100,55	101,81	94,32
Si	3,25	3,19	3,24	3,19	3,24	3,23	3,25	3,25
Al <sub>IV</sub>	0,75	0,81	0,76	0,81	0,76	0,77	0,75	0,75
Al <sup>VI</sup>	1,79	1,89	1,87	1,86	1,85	1,88	1,86	1,80
Ti	0,01	0,00	0,00	0,00	0,00	0,00	0,01	0,01
Fe <sup>2+</sup>	0,13	0,05	0,08	0,06	0,10	0,09	0,08	0,05
Mg	0,09	0,06	0,06	0,08	-0,05	0,05	0,08	0,15
Mn	0,00	0,00	0,00	0,00	0,00	0,00	0,00	0,00
Ca	0,00	0,00	0,00	0,00	0,01	0,00	0,00	0,01
Na	0,01	0,02	0,01	0,01	0,01	0,01	0,01	0,01
K	0,74	0,81	0,73	0,83	0,76	0,75	0,73	0,84
Inter Charg	0,75	0,83	0,74	0,84	0,78	0,76	0,74	0,86
∑oct	2,01	2,01	2,02	2,01	2,00	2,01	2,02	2,00
Fe/(Fe+Mg)	0,58	0,65	0,57	0,59	0,76	0,63	0,52	0,58

#### Table 6

			Epidote						Allanite				
$\overline{Na_2O(Wt\%)}$	0,10	0,15	0,10	0,04	0,00	0,06	0,04	0,10	0,04	0,03	0,02	0,08	0,04
MgO	0,00	0,01	0,11	0,01	0,18	0,22	0,16	0,12	0,44	0,28	0,80	0,12	0,25
$Al_2O_3$	19,84	21,31	19,26	21,06	22,08	13,59	13,69	13,65	13,19	14,56	17,23	15,82	18,27
SiO <sub>2</sub>	34,15	36,96	32,96	35,67	37,94	28,98	30,00	29,06	28,79	28,03	32,56	28,11	33,05
K <sub>2</sub> O	0,05	0,11	0,03	0,04	0,01	0,08	0,04	0,06	0,10	0,03	0,03	0,06	0,03
CaO	20,28	21,00	20,58	21,65	21,46	10,15	10,74	10,64	12,00	10,45	11,03	11,11	11,30
TiO <sub>2</sub>	0,11	0,07	0,04	0,04	0,45	0,23	0,44	0,47	1,70	0,22	0,64	0,24	0,13
MnO	0,00	0,06	0,02	0,12	14,74	0,20	0,33	0,13	0,14	0,22	0,24	0,48	0,62
FeOt	14,34	14,39	14,68	14,23	0,01	15,25	15,48	14,41	14,45	12,26	12,40	10,76	12,01
$Nd_2O_3$	0,00	0,00	0,16	0,00	0,36	3,07	1,79	1,03	1,34	3,15	3,25	1,48	2,83
$Ce_2O_3$	0,32	0,15	0,20	0,90	0,00	10,11	9,82	9,55	9,87	10,91	10,37	9,84	9,23
$La_2O_3$	0,38	0,13	0,27	0,00	0,09	6,67	6,04	5,80	4,41	5,92	5,70	5,52	5,40
Total	89,57	94,33	88,42	93,76	97,32	88,61	88,56	85,03	86,46	86,06	94,26	83,64	93,16
Si	2,99	3,05	2,95	2,99	3,03	2,95	3,00	3,01	2,94	2,93	3,01	2,95	3,05
Al	2,05	2,07	2,03	2,08	2,08	1,63	1,62	1,67	1,59	1,79	1,88	1,96	1,99
Ti	0,01	0,00	0,00	0,00	0,03	0,02	0,03	0,04	0,13	0,02	0,04	0,02	0,01
Mn	0,00	0,00	0,00	0,01	0,00	0,02	0,03	0,01	0,01	0,02	0,02	0,04	0,05
Fe <sup>2+</sup>	0,94	0,89	0,99	0,90	0,89	1,17	1,17	1,12	1,11	0,97	0,86	0,85	0,83
Mg	0,00	0,00	0,01	0,00	0,02	0,03	0,02	0,02	0,07	0,04	0,11	0,02	0,03
Ca	1,90	1,86	1,98	1,94	1,84	1,11	1,15	1,18	1,31	1,17	1,09	1,25	1,12
Na	0,02	0,02	0,02	0,01	0,00	0,01	0,01	0,02	0,01	0,01	0,00	0,02	0,01
K	0,01	0,01	0,00	0,00	0,00	0,01	0,00	0,01	0,01	0,00	0,00	0,01	0,00
Nd	0,00	0,00	0,01	0,00	0,01	0,16	0,09	0,06	0,07	0,17	0,15	0,08	0,13
Ce	0,01	0,00	0,01	0,03	0,00	0,38	0,36	0,36	0,37	0,42	0,35	0,38	0,31
La	0,01	0,00	0,01	0,00	0,00	0,25	0,22	0,22	0,17	0,23	0,19	0,21	0,18
Fe/(Fe+Al)	0,32	0,30	0,33	0,30	0,30	0,42	0,42	0,40	0,41	0,35	0,31	0,30	0,30

#### Table 7

		Monazite (K	A6 437,85)			Monazite (0	GR1 631)	
SiO <sub>2</sub> (Wt %)	0,37	1,22	0,39	0,43	0,88	0,22	0,72	0,42
$P_2O_5$	30,73	29,76	30,99	30,76	30,13	28,66	29,55	29,66
CaO	1,14	1,37	1,12	1,18	0,28	0,25	0,28	0,20
$Y_2O_3$	0,97	0,90	0,99	1,04	0,74	0,79	0,68	0,59
$La_2O_3$	15,05	14,98	15,61	15,63	18,74	18,17	17,30	20,10
$Ce_2O_3$	30,47	29,80	30,31	30,07	33,50	33,26	33,76	33,16
$Pr_2O_3$	2,99	2,92	2,95	3,04	3,54	3,88	3,33	3,21
$Nd_2O_3$	13,10	11,93	12,42	12,01	8,20	8,76	8,30	7,32
$Sm_2O_3$	1,40	1,65	1,51	1,65	0,59	0,67	0,37	0,59
$Gd_2O_3$	0,72	0,69	0,63	0,79	0,37	0,79	0,47	0,08
PbO	0,05	0,13	0,08	0,09	0,02	0,02	0,04	0,00
ThO <sub>2</sub>	0,87	1,33	0,72	0,96	0,03	0,00	0,01	0,01
UO <sub>2</sub>	0,01	0,01	0,00	0,02	0,01	0,09	0,04	0,01
Total	97,46	96,27	97,29	97,22	97,04	95,57	94,85	95,36
Th/La	0,06	0,09	0,05	0,06	0,00	0,00	0,00	0,00
ΣREE	64,71	62,86	64,42	64,23	65,68	66,32	64,20	65,07

					2 sigma		2 sigma			2 sigma
					error		error		Age (Ma)	error
				Pb207/U23	Pb207/U23	Pb206/U23	Pb206/U23		Pb207/Pb2	Pb207/Pb2
Pb ppm	Th ppm	U ppm	Th/U	5	5	8	8	Rho	06	06
473	5291	42	125	10,43	0,51	0,35	0,01	0,76	2949	76
525	6228	56	112	7,68	0,33	0,26	0,01	0,82	2956	64
370	4400	34	131	10,04	0,56	0,32	0,01	0,72	3031	89
341	3703	29	126	10,77	0,55	0,35	0,01	0,75	3001	79
415	4816	45	107	9,56	0,46	0,31	0,01	0,77	2997	74
580	6964	60	116	10,14	0,45	0,33	0,01	0,81	3026	67
507	5934	59	100	8,48	0,44	0,27	0,01	0,74	3016	82
316	2877	32	89	9,73	0,45	0,31	0,01	0,79	3032	71
384	4442	48	92	7,21	0,41	0,24	0,01	0,70	2993	91
			-							-
558	7279	65	112	5,43	0,44	0,22	0,01	0,58	2640	138
786	10178	125	82	3,43	0,21	0,14	0,01	0,65	2598	102
861	10420	84	124	6,10	0,28	0,25	0,01	0,77	2610	73
680	7568	60	126	6,95	0,33	0,29	0,01	0,76	2618	75
740	8807	81	109	6,24	0,28	0,27	0,01	0,77	2542	72
303	3611	50	72	4,15	0,25	0,17	0,01	0,65	2622	100
509	7707	138	56	3,29	0,17	0,14	0,01	0,72	2559	82
273	3438	42	81	4,97	0,30	0,20	0,01	0,66	2658	99
296	3850	66	59	4,34	0,24	0,18	0,01	0,68	2648	92
345	4850	82	59	3,37	0,19	0,14	0,01	0,67	2599	93
676	7159	61	118	9,49	0,44	0,40	0,01	0,78	2584	72
391	3131	23	136	13,20	0,68	0,52	0.02	0.73	2698	82
283	1741	24	72	9,19	0,58	0,37	0.02	0,65	2642	104
798	9406	68	139	8.30	0.37	0.32	0.01	0.80	2705	68
					,	,	,	,		
748	11219	117	96	5,68	0,24	0,18	0,01	0,83	3049	63
738	11372	122	93	4,90	0,23	0,17	0.01	0,77	2924	74
735	10807	118	91	4,97	0,22	0,17	0,01	0,80	2878	67
780	11614	99	117	5,89	0,26	0,20	0.01	0,80	2916	67
669	8923	63	142	8.68	0.39	0.29	0.01	0.80	2975	67
631	8724	66	132	7,70	0,33	0,25	0.01	0,81	3009	65
631	8254	56	146	7.38	0,36	0,27	0.01	0,75	2841	77
736	9649	62	157	7,75	0,35	0,26	0,01	0,79	2930	69
597	7892	52	152	6,15	0,31	0,21	0.01	0,74	2942	80
490	6660	56	119	4,86	0,26	0,16	0.01	0,72	2936	84
707	9391	60	156	6,69	0,31	0,23	0.01	0,78	2920	71
754	10964	88	125	6.61	0.30	0.25	0.01	0.79	2780	69
680	9151	81	114	5,40	0,26	0,23	0,01	0,73	2576	78
736	9140	61	151	8,32	0,38	0,32	0,01	0,78	2733	71
622	7734	60	128	6,51	0,32	0,23	0,01	0,74	2891	78
450	5431	38	145	10.28	0,49	0,36	0,01	0,76	2865	73
555	6692	50	135	10.38	0,48	0,36	0,01	0.77	2912	72
606	7235	40	180	10.74	0,52	0,37	0,01	0,76	2927	75
556	6780	64	106	8,36	0,37	0,28	0,01	0,79	2941	67
352	2909	36	81	7,49	0,46	0,30	0,01	0,65	2658	101

GR1-631						2 sigma error		2 sigma error		Age (Ma)	2 sigma error
Analysis	Ph nnm	Thoom	I nnm	Th/∐	Pb207/I1235	Pb207/11235	Ph206/11238	Ph206/U238	Pho	Ph207/Ph206	Ph207/Ph206
05120912c	231	1429	337	4.2	5.78	0.24	0.35	0.01	0.85	1976	63
06120912c	143	582	316	1,8	5,38	0,24	0,33	0,01	0,78	1931	73
07120912c	265	1799	399	4,5	5,51	0,22	0,33	0,01	0,87	1948	60
08120912c	191	1387	199	7,0	7,69	0,35	0,36	0,01	0,78	2401	70
17120912c	72	31	224	0,1	6,03	0,31	0,35	0,01	0,68	2037	86
18120912c	125	442	264	1,7	6,49	0,36	0,35	0,01	0,65	2157	92
19120912c	96 159	224	303	0,3	5,87	0,32	0,34	0,01	0,66	2057	90
20120912c	158	234	400	0,5	5,78	0,31	0,34	0,01	0,66	1993	90 84
21120912c	322	25	294	0,1 7 7	10,97	0,00	0,38	0.01	0,00	2760	88
23120912c	363	1934	359	5.4	14.02	0,50	0,38	0.01	0,64	3161	86

GR5-510       11,5       -5,33         GR23-630,9       10,5       -13,76         GR23-631,5       10,9       -9,56         GR43-636       10,4       -7,02         GR43-638       10,2       -6,80         KA6-462       17,7       -10,65	GR5 510	$\delta^{18}O$ (‰ SMOW)	$\delta^{13}$ C (‰ PDB)	7
GR23-630.9       10.5       -13.76         GR23-631.5       10.9       -9.56         GR43-636       10.4       -7.02         GR43-638       10.2       -6.80         KA6-462       17.7       -10.65	NINJ-J1U	11.5	-5.33	1
GR23-631,5 10,9 -9,56 GR43-636 10,4 -7,02 GR43-638 10,2 -6,80 KA6-462 17,7 -10,65	GR23-630,9	10,5	-13,76	1
GR43-636       10,4       -7,02         GR43-638       10,2       -6,80         KA6-462       17,7       -10,65	GR23-631,5	10,9	-9,56	1
	GR43-636	10,4	-7,02	
	GR43-638	10,2	-6,80	
	KA6-462	17,7	-10,65	