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Accurate REE measurement by ICP-MS after ion exchange separation. Application to ultra-depleted samples.

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<u>Abstract</u>

This study reports precise and accurate data for rare earth elements (REE) measured on eight geological reference materials, five enriched in REE (BE-N, BHVO-2, BR, BR-24 and RGM-1) and three very depleted in REE (BIR-1, UB-N and DTS-2). Data were acquired by quadrupole ICP-MS after isolation of the REE using an ion-exchange chromatography procedure. All the measured REE abundances are similar within $\approx 5\%$ (10% for the most REE-depleted sample DTS-2) to the high quality measurements previously published in the literature. We also show that using an internal Tm spike, the reproducibility of the data is improved to ~1%.

Applying this technique to the analysis of ultra-depleted rock samples (sub ng.g⁻¹), we show that significant improvements are obtained relative to the routine trace element measurement method. The chondrite-normalized patterns are smooth instead of displaying irregularities. Although the classical method gives excellent results on REE-rich samples, we believe that our technique improves the precision and accuracy of measurements for highly REE-depleted rocks.

Introduction

For decades, most geochemical studies use trace element data to identify geological processes such as melting of mantle sources or fractional crystallization of magmas (Henderson, 1984). In particular, the accurate determination of Rare Earth Element (REE) concentrations provides important information for the understanding of geological processes. Because of their high charge (trivalent cations except for Ce⁴⁺ under oxidizing conditions and Eu²⁺ under reducing conditions) and large radii, REE are incompatible elements during most mantle melting processes (White, 2003), and they are usually more abundant in melt-derived rocks than in residual ultramafic rocks. The extreme REE-depletion currently observed in the latter type of samples (sub ng.g⁻¹) explains why it is so difficult to measure them accurately (e.g. Nakamura and Chang, 2007).

Quadrupole inductively coupled plasma-mass spectrometer (ICP-MS) is one of the most commonly used technique to quickly and precisely determine the REE concentrations,

even at sub μ g.g⁻¹ concentrations in rock samples. Nevertheless, when the concentrations are lower, at the sub ng.g⁻¹ level, measurements using quadrupole ICP-MS become problematic due to sensitivity limitation. For samples very depleted in REE, other methods might thus be preferred. Isotopic dilution coupled with thermal ionization mass spectrometry has a wellestablished reputation as it provides very precise data (i.e., Raczek et al., 2001). However, this method is time-consuming and only elements with more than one isotope can be measured. Other alternatives exist: for example, Willbold and Jochum (2005) use isotope dilution coupled to a sector field ICP-MS and Jain et al. (2000) use an ultrasonic desolvating nebulisation coupled to ICP-MS. These techniques provide very good data but they either require specific instruments or are significantly more time consuming than the usual ICP-MS analysis.

Nakamura and Chang (2007) reported recently precise REE data on highly depleted reference materials (PCC-1, DTS-1, DTS-2 and JP-1) using a quadrupole ICP-MS and appropriate mathematical corrections for oxide interferences. Although this method provides quite reproducible data ($\approx 5\%$, 1 σ), it is also possible to enhance the signal and reduce the oxide interferences by isolating the REE from the other elements present in the rock sample (Hirata et al. 1988, Barrat et al. 1996).

Here, we describe a procedure including chemical separation of the REE and measurement by quadrupole ICP-MS. Based on this procedure, we report results for REE contents in eight geological reference materials (BE-N, BHVO-2, BR, BR-24, RGM-1, BIR-1, UB-N and DTS-2) as well as few ultra-depleted peridotites from New Caledonia.

Analytical method

Chemical isolation of Rare Earth Elements

The protocol described in this paper is adapted from the methods published by Strelow (1966) and Barrat et al. (1996). All procedures were carried out under cleanroom conditions, all acids (HCl, HNO₃, HF and HClO₄) were double distilled and de-ionized water (resistivity of 18.2 M Ω cm⁻¹) was used throughout the protocol.

About 50 mg of sample powder is dissolved in a mixture of concentrated HF and HCLO₄ (5:1) in Teflon containers maintained in steel jacket PARR bombs at 140°C for five days to achieve complete dissolution. Samples are then transferred in clean Savillex teflon

beakers and placed on a hot plate at 150°C until the HF-HCLO₄ mixture is completely evaporated. The residue is taken up in 3 mL of 6N HCl and a variable quantity (depending on the estimated REE concentration in the sample) of a Tm solution is added to samples in order to have a final Tm spike concentration of ~15 ppb (see supplementary table 1). The sample is heated at 100°C for 24 hours and finally evaporated. The dry residues are taken up in 2 mL of a mixture of 7N HNO₃-6N HCl (3:1), fluxed for 12 hours, ultrasounded for 10 minutes and centrifuged for 5 minutes at 5000 rpm to verify the absence of solid residue. If necessary, these latter crucial steps are repeated because the presence of a solid residue could entrain a loss of REE.

The 2 mL solution is loaded on a column (Biorad Poly-prep® columns) packed with 2 mL of 200-400 mesh Biorad® AG50W-X8 cation resin and calibrated to isolate REE from most other elements present in rocks. The resin is conditioned with 10 mL of a 7N HNO₃ - 6N HCl (3:1) mixture. All elements except REE are removed using 8 mL (1+1+6 mL) of HNO₃-HCl and REE are collected with 15 mL of 7N HNO₃. A typical elution profile is shown in Figure 1 where we report counts for REE and other trace elements as measured in 2 mL acid fractions of an elution performed using a UB-N dissolution. Samples are then evaporated to dryness and a mixture of 1 mL 14N HNO₃ and 0.5 mL 30% H₂O₂ is added to the residue to destroy any resin potentially present in the beaker and finally dried down. Just before analysis on the ICP-MS, samples are taken up in a weighted quantity of 2% HNO₃ that is adjusted to the required dilution factor (see supplementary table 1).

Instrumental, acquisition time and wash cycle

Measurements were carried out using an Agilent 7500ce quadrupole ICP-MS (Agilent Technologies). Samples were introduced with a quartz Micromist-type nebulizer with a quartz spray chamber cooled at 2°C. The operating conditions are optimized for maximum sensivity on ⁷Li, ⁸⁹Y and ²⁰⁵Tl (typically 2 Mcps/ppm, 5 Mcps/ppm and 2.5 Mcps/ppm, respectively). The complete operating conditions are listed in Table 1. Blank contribution and oxide production data are summarized in supplementary table 2 available on the GGR website.

Comparatively to light REE, heavy REE are generally less abundant in samples and thus counting statistics dictate that a longer acquisition time improves errors on the measurements. Therefore, two different acquisition times were chosen: a acquisition time of 0.60 second per mass for masses ranging from 137 (Ba) to 163 (Dy), and a longer acquisition

time of 1.20 second per mass for the heavier REE (¹⁶⁵Ho to ¹⁷⁵Lu). The total acquisition time per sample is estimated at 57.5 seconds.

The wash cycle between samples was optimized to ensure complete washout and equilibration of the entire system. It includes a 10 seconds wash with de-ionized water followed by a 120 seconds wash with 5% HNO₃ and a 240 seconds wash with 2% HNO₃. These washout times were chosen so that the number of counts at the end of the washing cycle was similar to those measured during the first wash cycle, before any sample was measured.

Calibration of the signal

Calibration is performed using one blank solution and two different dilutions of the USGS natural reference material BHVO-2 that followed the same chemical separation as the samples. The set of REE concentrations used for BHVO-2 are those published by Eggins et al. (1997), but BHVO-2 was also measured as unknown: in this specific case, the signal was calibrated using the REE contents of BR published by Eggins et al. (1997) (see Table 2 for results). The choice of BHVO-2 as the best natural standard was dictated by two factors: (a) Chauvel et al. (in press) estimated that for an accurate determination of the trace element concentrations of samples as poor as BIR-1 or UB-N, it was better to calibrate the ICP-MS signal with a rock standard not too rich in trace elements; (b) even if the REE concentrations in BHVO-2 are not certified, BHVO-2 is one of the most analyzed geological reference material and its REE content is well constrained (fourth among the 100 most frequently requested reference materials on the GeoReM website; see GeoReM preferred values by Jochum and Nehring, 2006, and Jochum and Nohl, 2008). Finally, the blank and BHVO-2 solutions were analyzed every five samples during the entire sequence of measurements.

Interference and analytical drift corrections

Several studies showed that oxygen reacts with some elements to form oxides during analyzes on ICP-MS (e.g. Cheatham et al., 1993; Dulski, 1994; Aries et al., 2000; Newman et al., 2009). Oxides of the light REE (LREE) cause interferences on intermediate REE (MREE) and for accurate determination of the MREE, a correction is required. As described by Dulski (1994), we correct the oxide interferences using a correction factor calibrated daily with

single element solutions of Ba and Ce, and a solution of mixed Pr and Nd. The main oxide interferences observed during the course of this work are listed in supplementary table 2.

When REE analyses are performed on samples simply dissolved but not chemically separated, Ba oxide interferences are significant and known to alter the results obtained for Eu contents (Dulski 1994, Aries et al. 2000, Nakamura and Chang 2007). In this case, corrections can represent up to 70% of the peak measured on mass 151 for highly depleted samples. In our study, this interference is considerably reduced due to the removal, through chemical isolation of the REE, of a large proportion of the Ba present in the sample (> 80%) (Figure 1). Indeed, during the course of this study, the Ba oxide interference never represented more than 0.04% of the total counts measured on mass 151 (Eu) (values for the other corrected oxide interferences are reported in supplementary table 2). Thus, complex interference corrections such as those described by Nakamura and Chang (2007) are not required here. Finally, instrumental drift was corrected using the drift on the Tm spike.

Procedural blank and detection limits

To quantify the exogenous pollution during the analytical procedure, procedural blanks were prepared following the same procedure as used for rock samples. The average REE contents of 10 blanks performed over 1 year are given in supplementary Table 2. They range from 0.6 to 40 pg, values that are negligible relative to samples since the sample/blank ratio is at least higher than 100 for the most REE-depleted reference material DTS-2. Detection limits for all REE were determined as the concentration equivalent of three times the standard deviation of the procedural blank. Values expressed as rock equivalents are plotted in Figure 2 and are listed in supplementary Table 2. They range from ~1 to 120 pg.g⁻¹ values that compare favorably to values published in the literature (often 10 to 10000 pg.g⁻¹, see for example Eggins et al., 1997 and Willbold and Jochum, 2005). Our detection limits are thus sufficiently low relative to the REE concentrations of all the reference materials analyzed in this work, even for the ultra-depleted sample DTS-2 (Figure 2).

Results and discussion

The first analyzes were performed after chemical separation of the REE but without Tm spike addition. The REE signal was calibrated using two methods: pure REE solutions and a BHVO-2 solution. There is usually two different ways to calibrate the ICP-MS signal: multiple dilutions of synthetic solutions (e.g. Nakamura and Chang, 2010) and geological reference materials (e.g. Eggins et al., 1997). Synthetic solutions present the advantage of having certified concentrations, which should make the calculated results more accurate. In contrast, no geological reference material has certified values for its trace element contents. As a consequence, using as a reference for the signal calibration the element concentrations of geological reference materials make the analytical results dependent on the accuracy of the published concentrations. Nevertheless, Eggins et al. (1997) pointed out that using a natural reference material to calibrate the signal presents the advantage of sample/standard matrix matching. In this case, both calibration standard and samples are prepared the same way and they have similar major element compositions. In addition, all REE are present in natural proportions (i.e., higher abundances of the even elements relative to the odd elements), a situation that is not matched by the synthetic solutions.

In the initial stage of this study, we performed measurements of the geological reference material UB-N using these two calibration techniques. Figure 3 shows the deviation between our measured values and the values reported by Garbe-Schönberg (1993). Three dissolutions of UB-N were calibrated in two different ways: in the first case, we used commercial pure REE solutions (certified reference material CMS-1 with REE contents certified at 10 µg.g-1, Inorganic Ventures) and in the second case, we used BHVO-2 as reference material. Figure 3 shows clearly that the BHVO-2 based calibration gives results similar within \pm 5% to the values published by Garbe-Schönberg (1993), while those calculated using the REE solutions with different dilutions are always lower and strongly deviate from the published values, particularly so for the MREE. It is unclear what causes the large deviation observed when pure REE solutions are used to calibrate the signal. The lack of sample/standard matrix matching as highlighted by Eggins et al. (1997) may explain part of the discrepancy since the chemical separation performed here reduces the matrix effect but does not completely eliminate all non-REE (Figure 1). However, other more complex and not well understood effects obviously also contribute to the observed deviations, in particular those of the MREE.

Based on the experience shown in Figure 3, we calibrate the ICP-MS signal using the geological reference material BHVO-2. However, while calculated data for international rock material with very different REE contents were similar to published values for both LREE

and HREE, data for the MREE were occasionally lower than the literature values (~ 10%; Supplementary Table 3, Figure 4). We believe that this MREE depletion is due to an interference of organic material on the masses used for the measurement of the MREE. Indeed, during the REE elution, minute quantities of resin might fall into the beaker in which the REE fraction is collected. If the proportion of organic material present in the beaker varies between samples and between BHVO-2 and the samples run as unknown, the calculation made for the interfered masses is wrong and systematically biased. Addition of concentrated nitric acid and H₂O₂ to the sample after evaporation solved the problem, probably through destruction of organic molecules as visible with the noticed effervescence when H₂O₂ was added.

The final improvement to the measuring technique consisted in the systematic addition of Tm spike to all samples in order to better correct the signal drift during ICP-MS measurement. In Table 2, we report REE concentrations obtained with and without Tm spike addition for eight international geological reference materials (BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2, RGM-1 and UB-N). Results are also plotted as chondrite-normalized patterns in supplementary figure 1. The second column in Table 2 reports REE contents measured without performing a drift correction based on an internal standard while data reported in the forth column were obtained after addition of a Tm spike prior to chemical separation. In the latter case, the Tm spike was used to correct for drift through time during ICP-MS measurements. For each set of data, we calculated the external reproducibility expressed as the relative standard deviation in % (RSD) of the multiple independent determinations of each reference material (see Table 2).

As shown in supplementary figure 1, the REE data for all reference materials exhibit smooth patterns, generally consistent with the available literature values. However, the logarithmic scale used in this type of figure hides small differences and in Figures 5, 6 and 7, we choose to plot the deviations between our measurements and literature values to better evaluate the quality of the REE data obtained in this study. With the exception of the highly REE-depleted DTS-2, relative deviations for all REE in all samples are < 5% (Figures 5, 6, 7), and the RSD values for most REE are usually smaller than 5%, even without using the Tm spike addition technique. Nevertheless, comparison of data obtained with and without spike addition shows that measurements performed using Tm spike are more reproducible, with lower RSD at about 1% (Table 2).

Results obtained for the most REE-depleted samples (BIR-1, UB-N and DTS-2, Figure 7) are also remarkably similar to high-quality data previously published. More specifically, our measurements of the reference materials BIR-1 and UB-N using the Tm spike are within 5% of the previously published data (Garbe-Schönberg, 1993; Willbold and Jochum, 2005; Bayon et al., 2009). Eu and Tb values for the spiked UB-N are slightly lower than those published by Bayon et al. (2009), but are nevertheless consistent with the data published by Garbe-Schönberg (1993). For DTS-2, the most REE-depleted sample of all reference materials, few data have been published. Only Raczek et al. (2001) and Nakamura and Chang (2007) reported high-precision measurements for this sample, using ID-TIMS and ICP-MS respectively. When compared to these data, our results display differences generally smaller than 10% (see Figure 7 and supplementary figure 1). Considering the low REE content of this sample and the scarcity of published data, it seems that a ~10% difference is acceptable. Finally, the RSD values for DTS-2 are < 10% for the MREE and always < 5% for the other REE.

In summary, we believe that the most accurate and reproducible results are obtained (1) when the calibration is done with a geological reference material that follows the same chemical separation as the analyzed samples; (2) when concentrated nitric acid and H_2O_2 are added to the sample after chemical separation to destroy organic molecules coming from the resin; (3) when samples are spiked before chemical separation because it improves significantly the reproducibility of the analyses.

Application to highly REE-depleted samples: Example of the New Caledonia peridotites

Here we apply the method to the determination of REE contents in New Caledonia harzburgites. These rocks are characterized by extremely low REE concentrations, probably due to multiple melting events (Ulrich et al. 2010). Two samples previously analyzed by Ulrich et al. (2010) were selected and analyzed using the procedure presented in this study. Both are characterized by a U-shaped REE pattern similar to the reference material DTS-2. The REE concentrations are also similar to DTS-2, at the sub-µg.g⁻¹ to ng.g⁻¹ levels. Figure 8 show a comparison between results obtained using the classical method (i.e., analysis of all trace elements after rock dissolution but without chemical separation, as described by Chauvel et al., in press) and those obtained with the method developed in this study (data are available in the supplementary Table 4). Results obtained without chemical separation exhibit

irregular chondrite-normalized patterns attributed to the very weak signal during data acquisition and to the enhanced effect of interferences on minute peaks. In contrast, the REE data measured after chemical separation define smooth patterns, with little to no difference between spiked and not spiked samples (Figure 8). This illustrates the improvements brought by isolation of the REE for highly REE-depleted rocks. These results demonstrate that an improved protocol can provide more reliable data for highly depleted samples and contribute to a better interpretation of the REE patterns in terms of geological processes.

Conclusions

We provide here a method for the accurate and reproducible measurement of REE in geological samples, including extremely REE-depleted rocks. This includes powder sample dissolution, REE-separation using ion-exchange columns loaded with cation resin, and measurement with a quadrupole ICP-MS. This protocol, applied to eight magmatic reference materials (BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2, RGM-1 and UB-N) gives good external errors (< 5%), even for the most depleted samples (<10 %). The results are in good agreement with high quality data published in literature. We finally demonstrate that for highly REE-depleted samples (sub ng.g⁻¹), this procedure provides smooth chondrite-normalized patterns in contrast to the irregular patterns obtained using a classical analysis performed on whole rocks.

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Table and Supplementary Table captions

 Table 1: Instrumental operating conditions.

Table 2: Analytical results for geochemical reference materials BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2, RGM-1 and UB-N. Methods used in the various publications: Baker et al. (2002): ICP-MS; Bayon et al. (2009): ICP-MS; Carpentier et al. (2009): ICP-MS; Chauvel et al. (in press): ICP-MS; Eggins et al. (1997): ICP-MS; Garbe-Schönberg (1993): ICP-MS; Nakamura and Chang (2007): ICP-MS; Raczek et al. (2001): ID/TIMS and MIC-SSMS; Roy et al. (2007): ICP-MS; Ryder et al. (2006): HR-ICP-MS; Willbold and Jochum (2005): ID/SF-ICP-MS. RSD = Relative Standard Deviation.

Supplementary Table 1: Dilution factor and amount of spike used for the various reference materials.

Supplementary Table 2: Table providing the selected isotope for the analysis of each element, the total blanks, the detection limits and the corrected interferences. Values in parentheses correspond to the highest contribution of the oxide interference on the total counts measured on masses 151 (Eu), 157 (Gd), 159 (Tb) and 163 (Dy).

Supplementary Table 3: Results obtained for BIR 1, BR-24, RGM-1 and BHVO-2 without any H₂O₂ added before analysis on the ICP-MS.

Supplementary Table 4: Comparison between results obtained on two highly REE-depleted harzburgites from New Caledonia analyzed using our procedure (with and without spike addition) and using the classical trace-element protocol as described by Chauvel et al. (in press).

Figure Captions

Figure 1: Elution curves for the REE as well as few other trace elements as measured on a UB-N sample using an ion-exchange column loaded with 2 mL of 200-400 mesh Biorad® AG50W-X8 cation resin. The trace element contents were measured in 2 mL fractions all along the elution profile.

Figure 2: Detection limits plotted as sample equivalents and compared to the DTS-2 concentrations published by Nakamura and Chang (2007).

Figure 3: Deviation between the UB-N values published by Garbe-Schönberg (1993) and results obtained on three dissolutions of UB-N. The three measurements were calibrated using two different methods: (1) a calibration based on several dilutions of a commercial pure REE solution (certified reference material CMS-1 with REE contents at 10 μ g.g⁻¹, Inorganic Ventures) and (2) a BHVO-2 calibration.

Figure 4: MREE depletion as observed during the analysis of BIR-1, BR-24, RGM-1 and BHVO-2 when no H_2O_2 is added to destroy organic matter. Data are given in supplementary table 3.

Figure 5: Comparison of our measured values for BE-N, BR and BR-24 with values from the literature as listed in Table 2, for both spiked and not spiked samples. The grey area corresponds to a relative deviation of 5%. The error bars correspond to one standard deviation on duplicate analyses but in some cases, the error bar is smaller than the symbol.

Figure 6: Comparison of our measured values for BHVO-2 and RGM-1 with values from the literature as listed in Table 2, for both spiked and not spiked samples. The grey area corresponds to a relative deviation of 5%. The error bars correspond to one standard deviation on duplicate analyses but in some cases, the error bar is smaller than the symbol.

Figure 7: Comparison of our measured values for BIR-1, UB-N and DTS-2 with values from the literature as listed in Table 2. The grey area corresponds to a relative deviation of 5% but for DTS-2 the field is extended to 10%. The error bars correspond to one standard deviation on duplicate analyses but in some cases, the error bar is smaller than the symbol.

Figure 8: Comparison of the chondrite-normalized patterns measured on two highly REEdepleted harzbugites from New Caledonia analyzed using our protocol (with and without spike addition) and using the classical trace-element protocol described by Chauvel et al. (in press). The chondritic values used for the normalization are from Anders and Grevesse (1989).

Supplementary figure 1: Chondrite-normalized patterns for BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2, RGM-1 and UB-N as measured in this study. Data acquired by others and listed in Table 2 are plotted for comparison. The chondritic values are from Anders and Grevesse (1989).



Figure 1

Figure 2





Figure 4









Figure 6



Figure 7





Supplementary figure 1: Chondrite-normalized patterns for BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2, RGM-1 and UB-N measured in this study. Data acquired by others and listed in Table 2 are plotted for comparison. C1-Chondrite values are from Anders and Grevesse (1989).

Table 1: Instrumental operating conditions.			
Parameter	Value		
Instrument	Agilent 7500ce		
Plasma power	1550 W		
Reflected power	1-5 W		
Torch	Quartz glass torch 2.5mm with injector		
Sampling depth	8-9 mm		
Plasma cool gas flow	15 l.min ⁻¹		
Auxiliary gas flow	0.90 l.min ⁻¹		
Sample gas flow	1.00 l.min ⁻¹		
Carrier gas flow	0.85 l.min ⁻¹		
Makeup gas flow	0.15 l.min ⁻¹		
Nebuliser	Quartz Micromist type 1 µL.min ⁻¹ - 400 µL.min ⁻¹		
Spray chamber	Quartz spray chamber, cooled at 2 °C		
Sample uptake rate	0.12 ml.min ⁻¹		
Sampling orifice	1.0 mm (made of Ni)		
Skimming orifice	0.4 mm (made of Ni)		
Typical sensitivity	2 Mcps for ⁷ Li, 5 Mcps for ⁸⁹ Y and 2.5 Mcps for ²⁰⁵ TI		
Oxide ratio (156:140)	< 1 %		
Double charge (70:140)	< 3 %		
Acquisition mode	Spectrum (multi Tune)		
Samples per peak	3		
Integration time per mass	0.6 s from 137 Ba to 163 Dy; 1.20 s from 165 Ho to 175 Lu		
Number of scans	100		
Calibration	External		
Internal standard	¹⁶⁹ Tm		

Table 2: Analytical results for geochemical reference materials BE-N, BHVO-2, BIR-1, BR, BR-24, DTS-2,RGM-1 and UB-N.

BE-N							
		This	Refe	References			
Element	without spi	ke (n=4)	with spike	with spike (n=4)			
	Average (µg.g ⁻¹)	RSD (%)	Average (µg.g-1)	RSD (%)	press)	Baker et al. (2002)	
La	83.9	1.3	83.5	0.5	82.5	82.29	
Ce	154.0	1.3	155	0.1	155	152.3	
Pr	17.4	1.3	17.6	0.1	17.4	17.09	
Nd	66.2	2.1	66.6	0.6	67.7	65.98	
Sm	11.7	3.9	12.2	0.3	12.2	12.03	
Eu	3.57	4.1	3.67	1.1	3.67	3.619	
Gd	9.52	0.5	9.75	1.1	9.87	9.771	
Tb	1.21	3.4	1.25	0.7	1.26	-	
Dy	6.37	3.2	6.39	1.1	6.32	6.397	
Ho	1.06	1.8	1.09	0.3	1.09	-	
Er	2.57	2.0	2.56	2.1	2.59	2.572	
Tm	0.312	4.1	-	-	-	-	
Yb	1.81	4.5	1.88	0.6	1.84	1.771	
Lu	0.24	5.5	0.244	3.3	0.245	0.2411	

BR

		This	References			
Element	without spike (n=4)		with spike (n=5)		Fogins et al.	
	Average (µg.g ⁻¹)	RSD (%)	Average (µg.g ⁻¹)	RSD (%)	(1997)	Roy et al. (2002)
La	83.2	1.0	83.6	0.2	82.1	82.13
Ce	153.0	1.0	154.0	0.4	152	154
Pr	17.7	2.2	17.6	1.2	17.36	17.52
Nd	67.3	2.0	66.9	0.3	66.1	67.37
Sm	12.1	2.5	12.2	2.2	12.11	12.2
Eu	3.63	0.1	3.63	0.4	3.58	3.65
Gd	9.63	1.8	9.49	0.3	9.57	9.61
Tb	1.24	3.0	1.29	4.0	1.29	1.32
Dy	6.4	1.6	6.36	1.1	6.3	6.4
Но	1.08	0.9	1.08	0.9	1.087	1.09
Er	2.57	1.8	2.61	1.4	2.59	2.6
Tm	0.308	2.6	-	-	0.303	-
Yb	1.8	0.7	1.84	3.3	1.81	1.85
Lu	0.246	1.9	0.243	2.7	0.251	0.24

BR 24 This study References without spike (n=5) with spike (n=3) Carpentier et al. Element Chauvel et al. (2009) (2010) Average (µg.g-1) RSD (%) Average (µg.g-1) RSD (%) 33.6 La 33.4 3.4 34.2 0.3 33.7 Ce 72.7 3.2 74.3 0.3 73.9 73.9 Pr 9.53 3.2 9.67 0.5 9.56 9.61 Nd 39.8 3.3 40.2 0.5 39.8 39.9 Sm 8.29 5.1 8.33 0.7 8.34 8.36 Eu 2.54 5.0 2.56 1.1 2.51 2.53 Gd 7.48 5.2 7.37 1.8 7.33 7.28 Tb 1.03 3.2 1.09 3.1 1.10 1.03 2.3 5.77 Dy 5.80 5.77 0.6 5.69 1.05 1.6 1.06 0.1 1.06 1.05 Ho 2.68 2.0 Fr 2.76 0.9 2.73 2.7 Tm 0.349 1.8 2.14 2.2 2.11 2.2 2.13 2.13 Yb 0.297 0.298 1.4 0.31 0.297 Lu 1.8

Methods used in the various publications: Baker et al. (2002): ICP-MS; Bayon et al. (2009): ICP-MS; Carpentier et al. (2009): ICP-MS; Chauvel et al. (in press): ICP-MS; Eggins et al. (1997): ICP-MS; Garbe-Schönberg (1993): ICP-MS; Nakamura and Chang (2007): ICP-MS; Raczek et al. (2001): ID/TIMS and MIC-SSMS; Roy et al. (2007): ICP-MS; Ryder et al. (2006): HR-ICP-MS; Willbold and Jochum (2005): ID/SF-ICP-MS. RSD = Relative Standard Deviation.

Table 2 (continued)

RGM-1								
		Thi	s study		Refer	References		
Element	without spi	ike (n=4)	with spike (n=3)		Eggins et al.	Byder et al.		
	Average (µg.g ⁻¹)	RSD (%)	Average (µg.g ⁻¹)	RSD (%)	(1997)	(2006)		
La	23.5	0.4	23.2	1.8	23.2	23.1		
Ce	47.1	0.9	46.3	2.0	45.9	45.3		
Pr	5.47	1.4	5.34	0.9	5.32	5.33		
Nd	19.3	1.1	19.1	1.7	19.1	19.3		
Sm	3.82	2.8	3.88	2.2	3.94	3.96		
Eu	0.579	1.5	0.575	2.0	0.547	0.637		
Gd	3.78	1.7	3.54	4.8	3.56	3.66		
Tb	0.572	3.8	0.597	4.5	0.605	0.599		
Dy	3.67	1.9	3.62	0.1	3.6	3.65		
Ho	0.769	0.9	0.764	0.7	0.769	0.766		
Er	2.35	0.6	2.37	1.7	2.33	2.29		
Tm	0.364	0.7	-	-	-	0.37		
Yb	2.51	2.5	2.47	1.2	2.47	2.51		
Lu	0.393	0.9	0.383	1.0	0.386	0.388		

BHVO-2								
		This	study		Refer	References		
Element	without spike (n=4)		with spike (n=5)		Willbold and	Raczek et al.		
	Average (µg.g-1)	RSD (%)	Average (µg.g-1)	RSD (%)	Jochum (2005)	(2001)		
La	15.3	0.9	15.3	1.3	15.3	15.2		
Ce	37.8	1.1	37.8	1.1	37.6	37.5		
Pr	5.41	1.7	5.38	0.6	5.31	5.29		
Nd	24.6	1.3	24.5	0.3	24.5	24.5		
Sm	6.10	3.6	6.04	1.0	6.04	6.07		
Eu	2.06	3.0	2.06	0.5	2.05	2.07		
Gd	6.24	2.4	6.27	0.7	6.23	6.24		
Tb	0.922	1.7	0.92	0.7	0.933	0.936		
Dy	5.29	1.0	5.29	1.0	5.29	5.31		
Но	0.978	0.8	0.977	0.8	0.964	0.972		
Er	2.55	0.6	2.52	0.3	2.49	2.54		
Tm	0.332	1.2	-	-	0.321	0.341		
Yb	1.99	1.2	1.98	0.2	1.95	2.00		
Lu	0.275	1.1	0.269	1.4	0.269	0.274		

BIR-1									
		This	study		Refer	References			
Element	without spike (n=4)		with spike (n=3)		Willbold and	Bavon et al.			
	Average (µg.g ⁻¹)	RSD (%)	Average (µg.g ⁻¹)	RSD (%)	Jochum (2005)	(2009)			
La	-	-	0.595	0.9	0.604	0.600			
Ce	-	-	1.88	0.9	1.89	1.91			
Pr	-	-	0.374	0.3	0.374	0.372			
Nd	-	-	2.37	0.6	2.37	2.40			
Sm	-	-	1.11	1.6	1.09	1.102			
Eu	-	-	0.518	2.2	0.508	0.530			
Gd	-	-	1.84	2.0	1.79	1.81			
Tb	-	-	0.366	2.6	0.399	0.366			
Dy	-	-	2.55	0.6	2.52	2.59			
Но	-	-	0.571	0.6	0.559	0.591			
Er	-	-	1.74	0.8	1.68	1.74			
Tm	-	-	-	-	-	-			
Yb	-	-	1.63	1.0	1.62	1.63			
Lu	-	-	0.245	0.7	0.241	0.243			

Table 2 (continued)

UB-N								
		This	study		References			
Element	without spike (n=3)		with spike (n=5)		Garbe-Schönberg	Bavon et al.		
	Average (µg.g ⁻¹)	RSD (%)	Average (µg.g ⁻¹)	RSD (%)	(1993)	(2009)		
La	0.286	2.8	0.306	0.4	0.3	0.29		
Ce	0.754	0.6	0.770	0.5	0.8	0.77		
Pr	0.115	1.2	0.115	1.1	0.12	0.118		
Nd	0.579	1.3	0.594	1.1	0.60	0.613		
Sm	0.210	6.5	0.214	1.6	0.21	0.222		
Eu	0.078	9.5	0.080	1.9	0.08	0.087		
Gd	0.305	2.9	0.323	1.1	0.31	0.32		
Tb	0.058	0.4	0.059	0.9	0.06	0.063		
Dy	0.425	0.8	0.424	2.3	0.41	0.434		
Ho	0.090	1.0	0.094	1.0	0.09	0.099		
Er	0.284	5.3	0.287	0.4	0.28	0.299		
Tm	0.041	1.5	-	-	0.043	-		
Yb	0.295	0.6	0.295	0.5	0.28	0.299		
Lu	0.043	5.5	0.045	0.9	0.043	0.047		

DTS-2								
		This	study		References			
Element	without sp	oike (n=0)	with spike (n=5)		Raczek et al.	Nakamura et		
	Average (µg.g-1)	RSD (%)	Average (µg.g-1)	RSD (%)	(2001)	Chang (2007)		
La	-	-	0.0124	3.3	0.0127	0.0132		
Ce	-	-	0.0252	3.5	0.0254	0.0263		
Pr	-	-	0.0030	4.2	0.0032	0.0033		
Nd	-	-	0.0132	5.8	0.0131	0.0136		
Sm	-	-	0.0027	8.2	0.00302	0.0033		
Eu	-	-	0.0008	9.0	0.00087	0.0009		
Gd	-	-	0.0037	3.5	0.00304	0.0038		
Tb	-	-	0.0006	1.7	-	0.0006		
Dy	-	-	0.0044	4.1	0.00419	0.0047		
Но	-	-	0.0012	2.1	-	0.0013		
Er	-	-	0.0051	1.3	0.00465	0.0055		
Tm	-	-	-	-	-	0.0012		
Yb	-	-	0.01	0.1	0.00963	0.0107		
Lu	-	-	0.0021	3.2	0.002	0.0023		