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Florian Mathias Huber, Yury Totskiy, Remi Marsac, Dieter Schild, Ivan Pidchenko, et al.. Tc interaction with crystalline rock from Äspö (Sweden): Effect of in-situ rock redox capacity. Applied Geochemistry, 2017, 80, pp.90-101. 10.1016/j.apgeochem.2017.01.026 . insu-01715154

HAL Id: insu-01715154 https://insu.hal.science/insu-01715154

Submitted on 22 Feb 2018

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Accepted Manuscript

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PII: S0883-2927(16)30425-5

DOI: 10.1016/j.apgeochem.2017.01.026

Reference: AG 3837

To appear in: Applied Geochemistry

Received Date: 22 October 2016

Revised Date: 19 January 2017

Accepted Date: 22 January 2017

Please cite this article as: Huber, F.M., Totskiy, Y., Marsac, R., Schild, D., Pidchenko, I., Vitova, T., Kalmykov, S., Geckeis, H., Schäfer, T., Tc interaction with crystalline rock from Äspö (Sweden): Effect of in-situ rock redox capacity, *Applied Geochemistry* (2017), doi: 10.1016/j.apgeochem.2017.01.026.

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Tc interaction with crystalline rock from Äspö (Sweden): Effect of in-situ rock redox capacity Huber, Florian Mathias^{1*}, Totskiy, Yury¹, Marsac, Rémi¹, Schild, Dieter¹, Pidchenko, Ivan¹, Vitova, Tonya¹, Kalmykov, Stepan², Geckeis, Horst¹, Schäfer, Thorsten¹ ¹ Karlsruhe Institute of Technology (KIT), Institute for Nuclear Waste Disposal (INE), P.O. Box 3640, D-76021 Karlsruhe, Germany. ² Lomonosov Moscow State University (MSU), Department of Chemistry, Leninskie Gory 1-3, 119991, Moscow, Russia *corresponding author: florian.huber@kit.edu Karlsruhe Institute of Technology (KIT) Institute for Nuclear Waste Disposal (INE) Hermann-von-Helmholtz-Platz 1 76344 Eggenstein-Leopoldshafen Germany Phone: +49 721 60 82 2384 Fax: +49 721 60 82 3927

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28 Abstract

The interaction of Tc(VII) with crushed crystalline rock (Äspö diorite; 1-2 mm size fraction) 30 31 from the Äspö Hard Rock Laboratory (HRL) (Sweden) was studied by laboratory batch 32 sorption and desorption experiments under Ar atmosphere using both natural and synthetic 33 groundwater. The Äspö diorite used in the experiments was drilled, transported and handled 34 as far as possible under anoxic conditions to preserve the in-situ rock redox capacity. For 35 comparison, identical experiments using artificially oxidized Äspö diorite have been carried 36 out to examine the effect of in-situ redox capacity on Tc uptake. According to the batch 37 studies, Tc(VII) uptake on the Äspö diorite is strongly dependent on redox capacity. Uptake 38 on un-oxidized rock is approximately 2 times higher compared to oxidized rock samples, 39 most likely due to higher Fe(II) contents of the un-oxidized rock. Tc redox states and 40 speciation both on the mineral surface and in the bulk were studied using X-ray 41 photoelectron spectroscopy (XPS) and Tc K-edge X-ray absorption near edge structure 42 (XANES) spectroscopy. The spectroscopic results verify a Tc(VII) reduction to Tc(IV) at the 43 rock surface. Distribution coefficients (K_d) and surface normalized distribution coefficients 44 (K_a) were determined and compared to available literature data. The formation of a Tc 45 colloidal phase was not observed under the geochemical conditions prevailing in the 46 experimental studies. Desorption of Tc is very low under anoxic conditions, but after artificial 47 oxidation Tc mobility is strongly increased. The results of this work clearly highlight the effect of in-situ rock redox capacity on Tc retention. 48

49

50 1. Introduction

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52 The generally accepted concept of spent nuclear fuel (SNF) and high-level nuclear waste 53 (HLW) long-term storage is by disposal in deep geological formations at depths of around 54 250 – 1000 meters (IAEA, 2001). The repository host rock as part of a multi-barrier system 55 plays an important role as retention barrier for radionuclide migration. Thus, the selection of 56 the host rock formation with appropriate geochemical and hydrogeological properties is a 57 key challenge during nuclear waste repository siting and it requires comprehensive scientific 58 research. Crystalline rocks (e.g. granites and gneisses) are considered as potential host rock 59 formations for the deep geological disposal in several countries (e.g. Sweden, Finland, 60 Russia, Korea). The work described herein focuses on crystalline rock samples from the Äspö 61 Hard Rock Laboratory (HRL) (Sweden) which is a generic underground research laboratory 62 (URL) located on the Äspö island near Oskarshamn in southern Sweden dedicated to in-situ 63 studies of processes in crystalline formations concerning deep geological disposal of spent 64 nuclear fuel (SKB, 2011). Radionuclide transport depends strongly on the bedrock 65 hydrogeological and geochemical conditions (pH, Eh, and ionic strength) and is governed by 66 different immobilization-remobilization processes (Grambow, 2008). The most important retention processes to be considered in fractured crystalline rocks are sorption to rock 67 surfaces, redox reactions and matrix diffusion (Bodin et al., 2003; Neretnieks, 1980; Xu and 68 69 Wörman, 1999).

⁹⁹Tc is one of the main long-lived U and Pu fission products in SNF and is also generated by 70 71 medical laboratories and research institutions. Because of its relatively high fission yield (ca. 6%) and long half-life (2.1×10⁵ years) ⁹⁹Tc is considered as a radioactive component of HLW 72 73 with significant toxic relevance (Kratz and Lieser, 2013). Technetium mobility in natural 74 systems strongly depends on the redox state. The most stable Tc form under aerobic 75 atmosphere is the pertechnetate ion TcO₄, which is very soluble and behaves like a 76 conservative tracer under oxidizing conditions (Rard et al., 1999). In an early work by 77 Bondietti and Francis (1979) using a variety of natural rock materials considerable Tc 78 retention due to reduction of pertechnetate was observed in accordance with Eh/pH

conditions. The potential of the TcO_4 / TcO_2 couple was described with the equation (1) 80 (Meyer and Arnold, 1991):

81

$$E_0(TcO_4^-/TcO_2) = 0.738 - 0.0788 \times pH + 0.0197 \times \log[TcO_4^-]$$
(1)

82 Under aerobic conditions, reported K_d values of Tc on crystalline rocks are negligible, < 1 mL/g in Allard et al. (1979) (contact time 1 day) and < 0.1 - 8.6 mL/g in Videnska and 83 84 Havlova (2012) (contact time with granitic rocks up to 94 days), whereas under reducing 85 conditions the values are much higher (50 mL/g in (Allard et al., 1979)). Both batch type 86 sorption and column experiments with Hanford sediments (Um and Serne, 2005; Zachara et al., 2007) have revealed that ⁹⁹Tc is highly mobile and shows virtually no retardation under 87 88 fully oxidizing conditions. However, under anoxic conditions TcO₄ is prone to reduction to 89 Tc(IV) and the solubility is limited by the hydrous oxide solid phase $TcO_2 \cdot 1.6H_2O_{(s)}$ (Meyer et 90 al., 1991). As a consequence, distribution coefficients and apparent diffusion coefficients of 91 technetium reported in the literature on natural minerals are scarce and are rarely published 92 together with the respective pe and pH conditions. Tc redox kinetics strongly depend on the 93 availability of reactive Fe(II) in the host rock and the mineral association and speciation on 94 surface (surface complexed, precipitated, and ion exchangeable) (Fredrickson et al., 2009; 95 Heald et al., 2007; Jaisi et al., 2009; Peretyazhko et al., 2008a; Peretyazhko et al., 2008b; 96 Zachara et al., 2007). Äspö in-situ and laboratory migration studies (CHEMLAB-2) using Äspö 97 derived natural groundwater revealed ≈1% Tc recovery (after 254 days) of the injected 98 Tc(VII) mass (Kienzler et al., 2003; Kienzler et al., 2009). Batch type studies done in parallel revealed surface normalized distribution coefficients, $K_{a_{\!\scriptscriptstyle A}}$ values of $\approx\!2.1{\times}10^{\text{-3}}\,\text{m}$ for ^{99}Tc 99 100 (t_{contact} = 14 d), whereas altered material showed significantly lower values. These results 101 revealed contact/residence time dependent retardation and/or reduction processes. In all 102 studies mentioned above, though carried out under e.g. Ar atmosphere in glove boxes, 103 crystalline rock material used were exposed for a considerable time to air before the 104 experiment. This circumstance might be one of the most important drawbacks in all batch 105 sorption/desorption studies carried out not only on redox sensitive radionuclides but for 106 redox sensitive heavy metals in general. Due to the drastically increased effort in preserving 107 the natural in-situ rock redox capacity during drilling, transporting and storing of the rock

108 material, almost no studies are available using non-oxidized material. In consequence, the 109 distribution coefficients, sorption and reduction properties and behaviour published in 110 numerous studies could be biased and accompanied with high uncertainties. So far, only one 111 study on U interaction with the same un-oxidized Äspö diorite (ÄD) material (provided by 112 KIT-INE) as used in the present study is available (Schmeide et al., 2014). Therefore, the main 113 motivation of this work is to investigate the effect of in-situ rock redox capacity on 114 technetium sorption behaviour by conducting classical batch experiments using non-oxidized 115 and oxidized rock material, respectively.

116

117 **2. Materials and Methods**

118 **2.1. Radionuclides**

119 2.1.1. ⁹⁹Tc

All Tc batch experiments with Tc concentration $\ge 10^{-9}$ M have been carried out using a 99 Tc stock solution (13 mM NaTcO₄) produced at the Institute of Hot Chemistry, Nuclear Research Centre (former FZK, now KIT) (Karlsruhe, Germany). The solubility limit of Tc(IV) in a wide range of pH-Eh conditions covering most of the natural systems is about 4.4×10^{-9} M (Duro et al., 2006). The detection limit of low-background liquid scintillation counting (LSC) (Quantulus, PerkinElmer, Inc., LSC cocktail Ultima Gold) for 99 Tc measurement is $\approx 10^{-10}$ M.

126 2.1.2. ^{95m}Tc

For experiments with Tc concentrations lower than 10⁻⁹ M, ^{95m}Tc with a much shorter half-127 life (61 day) and a main gamma emission line at 204.1 keV was applied. The isotope was 128 129 produced by proton irradiation of natural Mo foil (50 µm thickness) containing the natural 130 isotopic composition at ZAG Zyklotron AG (Karlsruhe, Germany). After cooling the foil was 131 transported to the Institute for Nuclear Waste Disposal (KIT-INE) and processed to separate 132 technetium according to the technique of (Boyd et al., 1960). The foil was dissolved in a 133 mixture of concentrated H₂SO₄ and 30% H₂O₂ and afterwards was slowly neutralized with 134 saturated NaOH (up to alkaline pH). The obtained alkaline solution was passed through a 135 column filled with the anion exchanger Dowex 1x8 (100-200 mesh particle size) with a total

136 volume ca. 3 mL. The column was washed first with 20 mL 1 M K₂C₂O₄ to remove residues of 137 molybdate and after rinsing with 20 mL of MilliQ water pertechnetate was eluted with 30 mL 138 1 M HClO₄. The last fraction was collected into 2 mL vials, which were measured with γ spectrometry and samples with ca. 90% of ^{95m}Tc were combined and neutralized with 139 140 concentrated NaOH. The purification level was monitored with ICP-MS and γ -spectrometry. 141 The use of another column filled with Teva® Resin (Eichrom Technologies, LLC) prior to 142 Dowex significantly increased chemical purity of Tc. Technetium separation on Teva Resin 143 column was performed from \approx 1.5 M HNO₃ media and after washing the column with 2 M 144 HNO₃ Tc was eluted with 8 M HNO₃ according to the technique reported by (Tagami and 145 Uchida, 1999). A further separation step on the Dowex column results in the purification 146 from NO₃, which may act as undesired oxidizing agent, being absent in natural deep 147 geological anoxic groundwaters. The nitrate concentration was initially controlled with 148 nitrate test strips (Merck) and subsequent ion chromatography (IC) analysis. Detection limit of gamma spectrometric analysis of ^{95m}Tc using a 10 mL vial geometry and a high-purity 149 germanium (HPGe) semiconductor detector was estimated to be $\approx 10^{-14}$ - 10^{-15} M (three hours 150 151 measurement time) depending on the age of the stock solution.

152 **2.2.** Solid materials

153 2.2.1. Äspö diorite

154 Diorite is the dominating rock type in the Äspö area (Kornfält et al., 1997). The general 155 mineralogical characterization of ÄD is presented in Table 1. Fresh Äspö diorite was obtained 156 from a drilling campaign at the Äspö HRL (Sweden) in 2011 within the EU project CP CROCK. 157 Details of the sampling procedure and material characterization were originally published 158 within a CROCK S&T contribution (Schäfer et al., 2012). During the core drilling, special care 159 was taken to minimize the exposure of the solid material to air. Therefore, the drilling 160 procedure was carried out with a double tube technique preserving to the best possible 161 anoxic conditions. Natural Äspö groundwater ($Eh_{SHE} \approx -240$ mV) permanently bubbled with N₂ 162 was used as a drilling fluid. After a short visual inspection (max. 5 min) under tunnel 163 atmosphere the cores were directly transferred into a transparent LD-PE bag, which was 164 evacuated three times (\approx -0.4 bar) and purged with nitrogen gas before

165 **Table 1.** Petrographic characterization of Äspö diorite (Byegård et al., 1998; Kornfält et al.,
166 1997).

Material	Rock type	Mineralogical composition, %		Structural
				characteristic
Äspö diorite	Quartz	Plagioclase,	30-50	Porphyritic,
	monzodiorite/	Quartz,	10-25	medium-grained
	granodiorite	K-feldspar,	10-30	
		Biotite,	10-25	
		Epidote,	3-15	
		Amphibole (mainly hornblende	e), < 10	
		Muscovite, titanite, apatite, t	luorite,	
		zircon, magnetite		

167

168 welding. The same procedure was applied with an Al bag for the second confinement to 169 prevent oxidation during transportation to the KIT-INE laboratories (see Figure A.1, bottom). 170 At KIT-INE the cores were stored in a barrel under Ar atmosphere (\approx 1 bar overpressure). 171 Two Äspö diorite drill cores (borehole KA2368A-01, cores#1.32 and #1.33) were selected for 172 the experiments. Both cores were chosen because of their (maximal) distance to the tunnel 173 wall (13.04 – 13.52 m (core #1.32) and 13.52 – 14.00 m (core #1.33)) and their petrological 174 characterization (fresh Äspö diorite) during the drilling campaign (Figure A.1, top and 175 middle). For preparation of the crushed material, the cores were transferred into an Ar 176 glovebox equipped with a circular diamond saw and cut into small discs (0.5-1 cm in width). 177 These discs were then manually crushed with a hammer and separated into several size 178 fractions by sieving. For the sorption experiments the 1-2 mm size fraction in diameter was chosen. The N₂-BET surface area of 0.16 m²/g was measured for this size fraction. This un-179 180 oxidized crushed material was stored permanently in the glovebox under Ar atmosphere (≤ 1 181 ppm O₂). Part of this crushed diorite material was exposed to air for one week for artificial 182 surface oxidation to investigate the influence of sample preservation and preparation on Tc 183 uptake. The chemical composition of the rock material used was determined at the Institute

184 for Geosciences, Johannes Gutenberg University (Mainz, Germany) by X-Ray fluorescence 185 (XRF) spectrometry (spectrometer MagiXPRO, Philips) with a Rh anode operated at 3.2 kW). 186 In Table A.1 the XRF data for the material studied is compared to the oxidized Äspö diorite 187 used by Huber et al. (2012) and Huber et al. (2010) and to material from (Byegård et al., 188 1998). A typical granodiorite composition (Nockolds, 1954) is also added in Table A.1 for 189 comparison. The general composition of the ÄD is typical of granitic rocks with high amount of SiO₂ and Al₂O₃ (quartz and feldspar) (Nockolds, 1954). The Fe²⁺ content of the solid phases 190 191 was measured by cerimetric titration (cerate oximetry) with potentiometric end point 192 determination (Close et al., 1966). The new un-oxidized ÄD shows a much higher amount of ferrous iron (\approx 39% Fe²⁺/Fe_{tot} vs. \approx 27% in old ÄD samples) compared to the old oxidized ÄD. 193 194 This result gives confidence in the approach carried out to preserve the in-situ rock redox 195 capacity. In the work from (Byegård et al., 1998) no information on Fe(II) compounds were 196 reported. In order to estimate the cation exchangeable Fe(II) amount on the mineral 197 surfaces of the Äspö diorite, a method proposed by Heron et al. (1994) was applied using 198 10 mL 1 M CaCl₂ (pH = 7) in contact with 2 g of granite for 24 h. Afterwards, an aliquot was 199 taken for Fe(II) quantification by the ferrozine technique. The analytical procedure of the 200 ferrozine technique is described in Viollier et al. (2000). Additionally, Fe data on the new un-201 oxidized and old oxidized AD material determined by XRF are given in the Table A.1, showing 202 that the overall Fe(II) redox buffer is drastically reduced for the oxidized samples. The ion-203 exchangeable Fe(II) fraction for the un-oxidized ÄD was quantified at approx. 4–6 μg/g, 204 whereas for the oxidized samples lower values around 1–3 μ g/g for ÄD are obtained. The 205 rather high uncertainty in the measurements is attributed to the natural heterogeneity of 206 the rock material. The content of ion-exchangeable Fe(II) is about three orders of magnitude 207 lower than Fe(II) amount in the bulk obtained by XRF (see Table A1). However, the ratio 208 between Fe(II) content in fresh and oxidized materials of the order of three is similar for 209 both types of measurements.

210 **2.3.** Groundwater

211 Different types of groundwater have been used in the experimental program. Besides the 212 natural Äspö groundwater (ÄGW) a synthetic groundwater simulant (ÄGWS) has been 213 prepared to mimic the CROCK drilling site outflow groundwater composition (see Schäfer et

214 al. (2012)). All chemicals used in the preparation of the synthetic groundwater were of 215 analytical grade. Solutions were prepared with deionized Milli-Q water which was stripped 216 with Ar prior to use. ÄGWS has a comparable composition to natural Äspö groundwater 217 sampled in-situ from borehole KA3600-F-2 sampled in a Ar pre-flushed 50 L Teflon-coated Al-218 barrel at the CP-CROCK site (Heck and Schäfer, 2012). Chemical compositions of the ÄGWS 219 used and natural groundwater samples from Äspö and Grimsel (GGW) (glacial melting water 220 analogue with low ionic strength) are presented in Table 2. Relatively high Eh values of ÄGWS and GGW is explained by the low concentration of the redox couples (e.g. 221 222 Fe(II)/Fe(III)) in the solution, which cannot be measured using a Pt-electrode.

223 Table 2. Overview of the chemical compositions of the synthetic Äspö groundwater simulant (GWS),

	synth. Äspö	Synth. Äspö	Äspö GW	Grimsel GW
	GWS	GWS after	(KA-3600-F-2)	(MI-shear zone)
		122h contact		
		time	<u> </u>	
рН	8.0 <mark>± 0.05</mark>	8.0 <mark>± 0.05</mark>	7.8 <mark>± 0.05</mark>	9.67 <mark>± 0.05</mark>
Eh	390 <mark>± 50mV</mark>	n.m.	-240 <mark>± 50mV</mark>	320 <mark>± 50mV</mark>
[Mg ²⁺]	104 ± 1.0 mg/L	104.6 mg/L	69.4 mg/L	12.6 μg/L
[Ca ²⁺]	1109 ± 94 mg/L	1134 mg/L	1135 mg/L	5.3 μg/L
[K ⁺]	19.35 ± 3.86 mg/L	21.56 mg/L	10.5 mg/L	
[Li ⁺]	2.53 ± 0.04 mg/L	2.50 mg/L	6.0 mg/L	
[Fe ^{2+.3+}]	n.m.	n.m.	0.2 mg/L	< D.L.
[Mn ⁻]	2.32 ± 3.02μg/L	23.8 μg/L	0.338 mg/L	< D.L.
[Sr ²⁺]	19.68 ± 0.29 mg/L	20.14 mg/L	19.9 mg/L	182 μg/L
$[Cs^+]$	<d.l< td=""><td>< D.L</td><td></td><td>0.79 μg/L</td></d.l<>	< D.L		0.79 μg/L
[La ³⁺]	n.m.	n.m.		< D.L.
[U]	0.05 ± 0.01 μg/L	1.70 μg/L	0.11 μg/L	0.03 μg/L
[Th]	0.02 ± 0.01 μg/L	0.07 μg/L	< 0.01 µg/L	< 0.01 µg/L
[Al ³⁺]	182.8 ± 56.3 μg/L	439.6 μg/L	13.3 μg/L	42.9 μg/L
$[Na^+]$	1929 ± 29 mg/L	1905 mg/L	1894 mg/L	14.7 mg/L
[Cl ⁻]	4749 ± 145 mg/L	4895 mg/L	4999 mg/L	6.7 mg/L
[Si]	n.m.	n.m.	4.7 mg/L	5.6 mg/L
[SO4 ²⁻]	409 ± 5.0 mg/L	411.88 mg/L	394.4 mg/L	5.8 mg/L
[F ⁻]	1.97 ± 0.09 mg/L	1.98 mg/L	1.41 mg/L	6.3 mg/L
[Br⁻]	21.17 ± 0.37 mg/L	20.96 mg/L	23.2 mg/L	
[NO ₃ ⁻]	n.m.	n.m.	n.m.	< D.L.

224 Äspö groundwater and Grimsel groundwater, respectively.

		ACCEF	PTED MANUSC	RIPT	
	[HCO ₃]	n.m.	n.m.	n.m.	3.0 mg/L
	[B]	306.5 ± 212.5 μg/L	146.1 μg/L	885 μg/L	
225	n.m. – not	t measured: D.L. – detectio	on limit.		

226

227 **2.4. Batch experiments**

228 2.4.1. Sorption experiments

229 Before the start of the batch sorption kinetic experiments the crushed material was 230 contacted with relevant groundwater simulant (GWS) for one day and exchanging the water 231 five times. This step was conducted to remove any remaining fines/colloids from the sieving 232 procedure. In case of un-oxidized ÄD the last step was performed with natural ÄGW to 233 establish more realistic conditions. All batch experiments were carried out in 20 mL LSC vials 234 (HDPE, Zinsser[©]) inside an Ar glovebox with O_2 concentrations \leq 1 ppm at room temperature (20 ± 2° C). The solid-liquid ratio chosen was 2.00g of granitic rock and 8.0mL of 235 236 groundwater (250 g/L). Sample duplicates were prepared and kept closed during the 237 sorption experiments to prevent oxidation of Fe(II) at the mineral surfaces. All sorption experiments were conducted at pH 8.1 ± 0.05. Tc(VII) concentrations of 1x10⁻⁵M, 1x10⁻⁸M 238 and 1x10⁻⁹M were used in the experiments. For measurement of ⁹⁹Tc content in the sample 239 240 supernatant after the desired contact time 1 mL aliquots were taken and added to 10 mL of 241 Ultima Gold LSC cocktail for analysis with LSC. To differentiate between potentially formed 242 colloidal phases (e.g. Tc(IV) eigencolloids) and dissolved species a phase separation by 243 ultracentrifugation (Beckman Optima XL-90, 90,000 rpm, 694,000 × g) for 1 h was applied. 244 Redox potential of selected samples was measured in the Ar glovebox by using a Metrohm 245 (Ag/AgCl, KCl (3 M)) electrode. The measurements were performed directly in the sample 246 without phase separation. The Eh values were recorded every hour and then corrected for 247 the standard hydrogen potential (Eh always denotes corrected Eh_{SHE} in this work). The Eh 248 values were recorded after ca. 1 hour contact time. The redox potential measurements in 249 the sorption experiments samples were carried out after about two weeks and 1 month 250 contact time, respectively. Every sample was measured over a period of one day in the open 251 vial in the Ar glovebox (< 1 ppm O₂) to monitor the Eh evolution. A typical time dependent 252 Eh evolution is shown in Figure 1. The initial drop in Eh is likely due to the equilibration of

253 the electrode with the solution, whereas the continuous drift to more oxidizing potentials in 254 the later period might be explained to be a result of oxidation due to traces of oxygen in the 255 Ar glovebox (< 1 ppm O_2) that seems to be enough to compensate the redox capacity of the 256 sample within 24 hours. Therefore, the lowest Eh value detected was used as most 257 representative for the undisturbed rock/water system in the closed vials.



Figure 1: Typical evolution for Eh measurement in synthetic Äspö groundwater simulant with unoxidized diorite sample ($[Tc] = 10^{-10}$ M). Dashed line marks the Eh value chosen as the final one.

258 2.4.2. Desorption experiments

259 Subsequent to the sorption kinetic experiments, desorption kinetic studies have been 260 conducted. The Tc containing supernatant of the sorption experiment samples (samples with 261 3 months contact time in the sorption kinetic experiments) was removed and 8 mL of fresh 262 Tc-free groundwater added (natural ÄGW and Grimsel groundwater (glacial meltwater 263 analogue)). For each contact time, the supernatant was always completely removed, 264 analysed by LSC and substituted with fresh Tc-free groundwater of the same volume. A 265 subset of the samples was oxidized under air for one month after the sorption experiments. 266 For this, the initial Tc-containing liquid phase was removed and dry material in the vial was 267 exposed to air before the fresh Tc-free groundwater was added under aerobic atmosphere. 268 For this experiment ÄGWS was used instead of natural ÄGW to keep oxidizing conditions. 269 Desorption experiments cover a time range between a few seconds and 1 month contact 270 time.

271 **2.5.** X-ray spectroscopy

272 2.5.1. X-ray photoelectron spectroscopy (XPS)

273 To examine the Tc surface speciation X-ray photoelectron spectroscopy (XPS) was applied. 274 For XPS analysis small un-oxidized Äspö diorite fragments with unpolished faces after cutting by circular saw were contacted with 10^{-5} M Tc(VII) in ÄGWS for two months and washed by 275 276 MilliQ water for a few seconds to prevent salt precipitation directly before the XPS analysis. 277 All preparation and measurement steps were performed under Ar atmosphere. Transport of 278 the samples from the Ar glovebox to the XPS spectrometer under anoxic atmosphere was 279 achieved by using an O-ring sealed vacuum transfer vessel (PHI model 04-110). XPS 280 measurements were carried out with the XPS system PHI 5000 VersaProbe II (ULVAC-PHI 281 Inc.) equipped with a scanning microprobe X-ray source (monochromatic Al K_{α} (1486.6 eV) X-282 rays) in combination with an electron flood gun and a floating ion gun generating low energy 283 electrons (1.1 eV) and low energy argon ions (8 eV) for charge compensation at isolating 284 samples (dual beam technique), respectively. The angle between sample surface and 285 analyser was set to 45°. Survey scans were recorded with an X-ray source power of 12 W and 286 pass energy of 187.85 eV. Narrow scans of the elemental lines were recorded at 23.5 eV pass 287 energy. All spectra were charge referenced to C 1s (hydrocarbon) at 284.8 eV. Data analysis 288 was performed using ULVAC-PHI MultiPak program, version 9.5.

289 2.5.2. X-ray absorption spectroscopy (XAS)

290 XAS experiments were performed at the INE-Beamline at the ANKA 2.5 GeV synchrotron 291 radiation facility, Karlsruhe Institute of Technology (KIT), Karlsruhe, Germany. The detailed 292 description of the instrumental setup of the INE-beamline is presented in Rothe et al. (2012). 293 Tc K-edge (21,044 eV) X-ray absorption near edge structure (XANES) spectra were collected 294 in fluorescence mode using one element VITUS Vacuum-Silicon Drift Detector (SDD, Munich, 295 Germany). The uranium mineral meta-schoepite ($UO_3 \cdot nH_2O$) was measured simultaneously 296 with all samples and the Tc(IV) and Tc(VII) references. Pertechnetate solution with concentration of 10^{-2} M was taken as a Tc(VII) reference and TcO₂×1.6H₂O_(s) solid phase was 297 298 prepared by TcO₄⁻ reduction in the electrochemical cell and used as Tc(IV) reference. The U 299 L₂ (20,948 eV) edge XANES spectra were used for energy calibration. The set of samples with

Tc concentrations of $\approx 10^{-3}$ M contacted with crystalline rock materials was prepared and 300 301 mounted in an inert gas cell under argon atmosphere (see Table 3 for an overview of all XAS 302 samples and references measured). During the measurements, argon continually flowed 303 through the cell. Data reduction and normalization was performed with the ATHENA 304 program part of the IFEFFIT software package (Ravel and Newville, 2005).

Sample	[Tc] <i>,</i> M	Description
Tc(VII) reference	0.01	TcO₄ ⁻ solution
Tc(IV) reference	-	Solid TcO ₂ covered with supernatant
Tc on ÄD	0.001	Centrifuged suspension

305 Table 3. List of samples studied by XAS.

306 2.6. Thermodynamic modelling

307 Geochemical speciation calculations and sorption modelling were conducted with the 308 geochemical speciation code PHREEQC (version 2) (Parkhurst and Appelo, 1999). The SIT 309 database provided with PHREEQC is used, in which the thermodynamic constants for Tc 310 correspond to the ones selected by the NEA Thermochemical Database (Guillaumont et al., 311 2003). Pourbaix diagrams were calculated and plotted with "Geochemist's Workbench" 312 (version 8.0, Aqueous Solution LLC) code with the default database thermo.dat also modified 313 for Tc species in accordance with NEA Thermochemical Database (Guillaumont et al., 2003).

Under the applied experimental conditions TcO_4^- and $TcO(OH)_2$ are the only relevant 314 315 dissolved species for Tc(VII) and Tc(IV), respectively. The redox reaction is:

$$\frac{TcO_4^- + 4H^+ + 3e^-}{16} \rightleftharpoons TcO(OH)_2 + H_2O$$
(2)

$$K_{VII/IV} = \frac{[TcO(OH)_2]}{[TcO_4^{-}][H^+]^4[e^-]^3}$$
(3)

- where $K_{VII/IV}$ is the conditional constant of reaction (2) at a given ionic strength. From here 318
- 319 the pe (pe = -log a_{a} - = 16.9×Eh at 25°C) is calculated according to the following equation:

$$pe = \left(\log K_{VII/IV} + \log\left(\frac{[TcO_4^{-}]}{[TcO(OH)_2]}\right) - 4pH_c\right)/3$$

$$\tag{4}$$

where $pH_c = -\log [H^+]$. The pe corresponding to the 50/50% Tc(VII)/Tc(IV) borderline on a Pourbaix diagram pertaining only to the aqueous species (denoted pe_{aq} or Eh_{aq} for redox potential) is found as follows:

(5)

$$pe_{aq} = (\log K_{VII/IV} - 4pH_c)/3$$

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336

Sorption of Tc(IV) onto mineral surfaces can also be taken into account. Because this study is
restricted to pH ≈8 and low ionic strength for the sorption studies, no surface site protolysis
or electrostatics are taken into account. We have considered the following simple reaction
(6) to describe sorption onto the mineral surface:

$$\equiv \mathbf{S} + \mathbf{T}c\mathbf{O}_{4}^{-} + 4\mathbf{H}^{+} + 3\mathbf{e}^{-} \rightleftharpoons \equiv \mathbf{S}\mathbf{T}c\mathbf{O}(\mathbf{O}\mathbf{H})_{2} + \mathbf{H}_{2}\mathbf{O}$$
(6)

where \equiv S is a generic surface site and $[\equiv$ S] can be calculated using equation (7) with the following parameters: site density: 1 site/nm²; surface area: 0.16 m²/g; S/V: 250 g/L.

$$[\equiv S] = site \ density \times surface \ area \times (S/V)/N_A = 6.64 \times 10^{-5} \text{mol/L}$$
(7)

where N_A is the Avogadro constant. A similar surface complexation approach was proposed
by (Cui and Eriksen, 1996) for Tc uptake by Fe(II)-bearing minerals. According to the reaction
(6) the reaction constant is calculated as (8):

$$K_{VII/IV,surf} = \frac{[\equiv STcO(OH)_2]}{[\equiv S][TcO_4^{-}][H^+]^4[e^-]^3}$$
(8)

Tc(IV) uptake on mineral surfaces is high (Westsik Jr et al., 2014), so $[=STcO(OH)_2]$ is expected to be much larger than $[TcO(OH)_{2(aq)}]$. By contrast, Tc(VII) uptake on minerals can be neglected (Wildung et al., 2004). Using the surface complexation model, the Tc(VII)/Tc(IV) borderline on a Pourbaix diagram accounting for both processes in solution and at the mineral surfaces (denoted pe_{surf} or Eh_{surf}) is calculated as follows:

$$pe = \left(\log K_{VII/IV,surf} + \log\left(\frac{[TcO_4^{-}]}{[\equiv STcO(OH)_2]}\right) - 4pH_c + \log[\equiv S]\right)/3$$
(9)

$$pe_{surf} = \left(\log K_{VII/IV_{surf}} - 4pH_{c} + \log[\equiv S]\right)/3$$
(10)

In other words, pe_{surf} is the pe value for 50% Tc uptake. Equation (10) applies only for sufficiently low $[Tc]_{tot}$ or high E_h , i.e. where the precipitation of $TcO_2 \cdot 1.6H_2O_{(s)}$ does not occur. Note that, in a modelling study of plutonium (Pu) uptake on kaolinite, (Marsac et al., 2015a) calculated borderlines between two Pu oxidation states pertaining only to the speciation at the kaolinite surface. Such type of calculation cannot be made in our study because Tc(VII) uptake on minerals is unlikely to be quantifiable, in contrast to the relevant Pu oxidation states.

351 3. Results

342

343

352 3.1. Batch sorption studies

353 3.1.1. Sorption on Äspö diorite

354 In the following the term "sorption" implies the total Tc uptake by the solid phase 355 independent of the underlying process (sorption only or reductive sorption). Time 356 dependent sorption of different Tc(VII) concentrations onto oxidized and un-oxidized ÄD are given in Figure 2 for $[Tc]_{tot} = 10^{-5}$ (Figure 2a), 10^{-8} (Figure 2b) and 10^{-9} M (Figure 2c). From the 357 Tc sorption kinetic experiments, it is evident that Tc uptake on un-oxidized material is much 358 higher than the artificially oxidized one. For the sample series of 10⁻⁸ M and 10⁻⁹ M [Tc]_{tot} on 359 360 un-oxidized material plateau values close to 100% sorption are obtained (after 90 days 361 contact time), whereas during the same observation period in experiments with oxidized ÄD only \approx 40% are removed from solution. Reaching the plateau value for Tc uptake takes more 362 time at lower $[Tc]_{tot}$ concentration. In samples with 10^{-5} M Tc much less relative uptake is 363 364 observed showing a plateau level for sorbed Tc of around 20 – 25% for the un-oxidized 365 samples and $\approx 10\%$ for the oxidized samples. The steady state is reached after approximately 366 seven days which is much faster than observed for experiments with lower [Tc]_{tot}.



Figure 2. Tc sorption kinetics for different Tc concentrations in presence of oxidized and un-oxidized ÄD (pH 8, I = 0.2 M).

The formation of colloidal Tc phases (eigencolloids) examined by comparison of Tc concentration in ultracentrifuged to non-ultracentrifuged samples was not detectable within the uncertainty limits (\pm 5-10%). Either these colloidal phases are not formed or are not stable under the Äspö GWS conditions chosen (ionic strength \approx 0.2 M, pH 8).

371 Redox potential measurements of Tc containing ÄGWS after two weeks and 1 month 372 equilibration time with oxidized and un-oxidized ÄD for three different [Tc]_{tot} are shown in 373 Figure 3, together with Pourbaix diagrams calculated for the ÄGW composition. For the 374 oxidized ÄD material the redox potential (Eh_{SHE}) does not change significantly as a function 375 of Tc concentration and is within the range of +230 to +280 mV. However, for un-oxidized 376 ÄD material two trends were observed during the Eh measurements: (a) for low Tc concentration (up to 10^{8} M) the Eh value decreases with time from 14 days to one month 377 and (b) for the highest Tc concentration used (10^5 M) the Eh value was unchanged within 378 379 the analytical uncertainty. Based on the measured redox potentials in combination with 380 thermodynamic considerations the following conclusions can be drawn (at least after 2 weeks contact time): (i) for $[Tc] = 10^{-9}$ M Tc(VII) is reduced to a Tc(IV) species in solution 381 whereas (ii) for $[Tc] = 10^{-8}$ M Tc(IV) precipitates as a solid phase. In case of Tc = 10^{-5} M (iii) 382 Tc(VII) might not be fully reduced to Tc(IV) since the measured Eh_{SHE} values are above the 383 borderline. Nevertheless, a fraction of Tc(VII) might be reduced since the borderline 384 385 represents already 50%/50% Tc(VIII)/Tc(IV). In this case the reduced Tc precipitates as a Tc(IV) solid phase. 386



Figure 3. Pourbaix diagrams for ÄGWS containing different Tc concentrations: (a) 10⁻⁵ M, (b) 10⁻⁸ M and (c) 10⁻⁹ M. Data points are given for oxidized (open symbols) and un-oxidized (filled symbols) ÄD for a contact time of two weeks (squares) and one month (circles).

387 **3.2. Desorption studies**

388 3.2.1. Desorption experiments without artificial oxidation

389 Tc desorption was experimentally monitored over one month using sorption samples for 390 which Tc had been previously equilibrated with rock material for three months. Very low Tc 391 desorption was observed in all cases studied irrespective of the nature of the material 392 (oxidized or un-oxidized) used for sorption. For the case of AD samples two types of natural 393 groundwater were used, namely Äspö and Grimsel GWs (representing glacial melt water 394 composition with low ionic strength, see Table 2). Tc was detected in the liquid phase only for samples after contact with 10⁻⁵ M Tc ÄGWS (Figure 4). After 1 day contact time 395 desorption achieved values of up to 7% of the Tc amount sorbed after the uptake studies. 396 397 This level remained relatively stable up to 30 days of contact time. For lower Tc 398 concentrations (10⁻⁸ M and 10⁻⁹ M initial concentration for sorption experiments) no 399 desorption was detected with the detection limit of the analytical method.



Figure 4. Desorption of Tc sorbed to oxidized and un-oxidized ÄD material by ÄGWS after 1 month pre-oxidation under atmospheric conditions. Results of the experiment after artificial oxidization and contact with ÄGWS and GGW are additionally included in (a). Please note, that these data points have the unit "% desorbed".

400 3.2.2. Desorption experiments with artificial oxidation

Desorption kinetics for ÄD with different [Tc]_{initial} after sample pre-oxidation is given in Figure
4. Pre-oxidation of the ÄD samples under aerobic conditions for 1 month before addition of
ÄGWS changed the Tc desorption behaviour drastically (Figure 4). Both sorption experiments
with originally oxidized and un-oxidized materials revealed the same desorption behaviour.
Desorption is fast with the main part of technetium being released after a few seconds
contact time and after one day a concentration plateau value was reached.

- 407 **3.3.** Spectroscopic analyses
- 408 3.3.1. X-ray photoelectron spectroscopy (XPS)

409 XPS analysis of an un-oxidized ÄD disc fragment after contacting with 10^{-5} M Tc(VII) in ÄGWS 410 for2 months under Ar atmosphere in the glovebox revealed that Tc is associated to mafic 411 (dark) minerals only (see Figure A.2), whereas on felsic (light) minerals no Tc was found. 412 Figure 5 (left) shows the measured Tc 3d XPS spectra with the characteristic $3d_{3/2}$ and $3d_{5/2}$ 413 peak



Figure 5. (left) Tc 3d XPS narrow scan spectrum of an AD sample after contacting with Tc(VII)containing AGWS. The green (dashed) and blue (dotted) curves show the fits to the $3d_{3/2}$ and $3d_{5/2}$ XPS elemental lines, respectively, the red line represents the residuum. (right) Normalized Tc K-edge XANES spectra of a sample after sorption of Tc onto AD and of Tc(IV) and Tc(VII) references.

414 positions. Due to the rather low Tc concentration used in the sorption experiment on the AD 415 disc the spectra is quite noisy. Both Tc(VII) and Tc(IV) should have two 3d peaks $(3d_{5/2} \text{ and }$ 416 $3d_{3/2}$) for each oxidation state. Since there are only two peaks in the area of interest, the 417 conclusion is that there is only one Tc oxidation state present at the mineral surface, and the 418 position of Tc(IV) 3d_{5/2} (TcO₂) at 256.8 eV reference line is closer to the experimental data 419 (255.9 eV) than the Tc(VII) 3d_{5/2} (NaTcO₄) 259.7 eV reference line. It was therefore 420 concluded that Tc found on the AD surface is in the tetravalent oxidation state. Binding 421 energy reference lines for TcO_2 and $NaTcO_4$ were taken from (Wester et al., 1987).

422 3.3.2. X-ray absorption near-edge structure (XANES)

423 XANES analysis provides bulk information on the Tc oxidation states after interaction of 10^{-3} 424 M Tc with ÄD. Normalized Tc K-edge XANES spectra are presented in Figure 5 (right) for Tc 425 on ÄD together with Tc(IV) and Tc(VII) references. The Tc K-edge XANES spectra of the TcO₂ 426 and TcO₄⁻ reference materials have characteristic spectral features, which readily allow Tc 427 oxidation state characterization. For example, the spectrum of the Tc(VII) reference, where 428 Tc is surrounded by four oxygen atoms in tetrahedral conformation, exhibits a pre-edge 429 absorption resonance at about 21,050 eV generated by the 1s \rightarrow 5p/4d transition (Altmaier

et al., 2011b). Since Tc(IV) generally possesses octahedral coordination, this transition is
forbidden and the pre-edge feature cannot be seen in the spectrum of the Tc(IV) reference
compound. The overall shape and energy position of the rising absorption edge of the Tc Kedge XANES spectrum of Tc on Äspö diorite is similar to the TcO₂ reference without the preedge feature of Tc(VII). These fingerprint approach allows to identify Tc(IV) as dominating Tc
species on ÄD in line with the XPS results.

436 **4. Discussions**

437 **4.1. Tc uptake mechanisms**

438 The spectroscopic investigations described above together with the batch 439 sorption/desorption experimental results and the thermodynamic calculations demonstrate 440 that Tc retention in a reducing condition is always coupled to a reduction process from 441 Tc(VII) to Tc(IV). Tc(VII) shows no or only very weak sorption. Thus, the mechanism leading 442 to the removal of Tc from solution may be attributed to (i) a sorption/surface complexation 443 of a Tc(IV) species after reduction in solution by e.g. dissolved Fe(II) and/or (ii) precipitation 444 of poorly soluble TcO₂·1.6H₂O_(s) again due to Tc(VII) reduction in solution by e.g. dissolved 445 Fe(II) species. Especially in those experiments with the highest Tc concentration a reduction/precipitation process might occur, as the Tc(IV) solubility limit of 4.4×10⁻⁹ M (Duro 446 447 et al., 2006) should be significantly exceeded in case of reduction. A third possible 448 mechanism of Tc retention could be (iii) the Tc(VII) reduction by structurally bound or 449 surface bound Fe(II) species. A general scheme of the processes potentially involved into 450 Tc(VII) immobilization is shown in Figure 6.

451 Considering only Fe(II) as a potential reducing agent for Tc(VII)/Tc(IV) the transformation
452 follows according to equation (11):

453
$$TcO_{4}^{-} + 3Fe^{2+} + 4H^{+} \rightarrow TcO(OH)_{2} + 3Fe^{3+} + H_{2}O$$
(11)

In a first attempt we simply tried to assess whether the available amount of Fe(II) is sufficient to reduce added Tc(VII). The ion-exchangeable Fe(II) content was taken as an approximation for the readily "available Fe(II)" and decreases from $(2.2\pm0.5)\times10^{-5}$ M in unoxidized ÄD to $(9\pm5)\times10^{-6}$ M in oxidized ÄD. Based on the solid to liquid ratio chosen (250

g/L) these values give $(1.8\pm0.4)\times10^{-7}$ mol/vial Fe(II) for un-oxidized ÄD and $(7\pm4)\times10^{-8}$ 458 459 mol/vial Fe(II) for oxidized ÄD. Thus, it is obvious that the material extracted and transferred 460 under normal ambient atmospheric conditions (oxidized ÄD) possesses a lower redox capacity. From the comparison of the Fe(II) content $((7\pm4)\times10^{-8} \text{ mol/vial Fe(II)})$ with the total 461 amount of added Tc (8×10⁻⁸ mol/vial (10⁻⁵ M Tc), 8×10⁻¹¹ mol/vial (10⁻⁸ M Tc) and 8×10⁻¹² 462 mol/vial (10⁻⁹ M Tc)) and taking into account that three electrons are required to reduce 463 Tc(VII) to Tc(IV) oxidized ÄD still provides enough ion-exchangeable Fe(II) to reduce 464 29.1 \pm 16.7% of total Tc in the sample containing 10⁻⁵ M Tc, which is close to the experimental 465 observation of 12±1% Tc being reduced. Furthermore, un-oxidized ÄD does not provide 466 enough Fe(II) for the complete reduction of 10^{-5} M Tc. These estimations are supported by 467 the finding that Eh values in the granite/water suspensions increase significantly when 10⁻⁵ 468



Figure 6. General scheme of Tc(VII) sorption and reduction processes.

469 M Tc(VII) are added. This amount is apparently sufficient to exceed the redox capacity of the 470 diorite material at a solid to liquid ratio of 250 g/L. In all batch experiments the granite 471 material should provide enough Fe(II) for the Tc(VII) reduction at lower [Tc]_{tot}.

472

473 **4.2.** Sorption parameters

- 474 The temporal change in Tc concentration during the batch sorption experiment can be
- 475 described with an exponential decay equation (13):

476	$C_t = \left(C_0 - C_{eq}\right)e^{-kt} + C_{eq}$	(13)
477	where C_0 and C_{eq} are the initial and equilibrium Tc concentrations (M), respectively,	and <u>k</u> is
478	the sorption rate coefficient (time ⁻¹). Hence, calculating sorption S_t as (14)	
479	$S_t = \left(1 - \frac{C_t}{C_0}\right) \times 100\%$	(14)
480	time dependent sorption data can be fitted with the rate equation (15):	

$$S_{e} = S_{eq} (1 - e^{-RE})$$

$$(15)$$

482 where S_t and S_{eq} are the sorption percentages at the moment t and at equilibrium (6 months 483 contact time), respectively. Sorption rate coefficients obtained from this fitting are 484 presented in Table 4. The uptake rates (k) are significantly higher for the experiments with 485 high total Tc concentration, which is a general observation in chemical reaction kinetics. A 486 closer inspection shows that fitting of experimental data obtained with highest Tc 487 concentrations to the kinetic rate model provides a better correlation when two exponential functions (rate constants) are taken. For instance, the sorption of 10⁻⁵ M Tc onto oxidized ÄD 488 489 can be fitted with two exponential functions with k values of 0.017±0.008 and 0.265±0.056 d⁻¹. This might be an indication of the existence of more than one retention mechanism, e.g. 490 surface sorption and precipitation, respectively. We have to emphasize, however, that the 491

Mate -rial	Fe(II) available, mg/g	Initial Tc concentration, mol/L	Eh, 1-2 months *, mV	k, d ⁻¹	K _d , L/kg	K _a , m	Tc sorbed after 6 months, %
ÄD		1.07×10 ⁻⁵	<mark>76</mark>	<mark>0.24±0.100</mark>	<mark>1.1±0.200</mark>	<mark>(6.80±0.80)×10⁻⁶</mark>	<mark>21±2.0</mark>
un-	4-6	<mark>(1.05±0.05)×10⁻⁸</mark>	<mark>-187</mark>	<mark>0.075±0.009</mark>	<mark>500±200</mark>	<mark>(2.90±1.30)×10⁻³</mark>	<mark>99.2±0.6</mark>
oxi- dized	Y	<mark>(1.10±0.10)×10⁻⁹</mark>	<mark>-142</mark>	<mark>0.036±0.004</mark>	<mark>900±800</mark>	(5.30 ±4.90)×10 ⁻ ³	99.2±0.6 99.5±6.0
ÄD	1 2	<mark>1.07×10⁻⁵</mark>	<mark>238</mark>	<mark>0.15±0.040</mark>	<mark>0.53±0.050</mark>	<mark>(3.30±0.30)×10⁻⁶</mark>	<mark>12±1.0</mark>
oxi-	1-2	<mark>(1.05±0.05)×10⁻⁸</mark>	<mark>280</mark>	<mark>0.017±0.010</mark>	<mark>3.6±1.000</mark>	<mark>(2.20±0.60)×10⁻⁵</mark>	<mark>47±8.0</mark>

492 **Table 4**. Main parameters obtained by Tc (VII) sorption experiments onto ÄD.

		ACC	EPTE	D MANUS	CRIPT			
	dized	(1.10±0.10)×10 ⁻⁹	<mark>264</mark>	<mark>0.007</mark> ±0.001	<mark>22±8</mark>	<mark>(1.40±0.5)×10⁻⁴</mark>	<mark>84±6.0</mark>	
493	* values ir	n Fig. 3; N/A – not appli	cable.					-

494

* values in Fig. 3; N/A – not applicable.

495 system consists of highly coupled redox, precipitation and surface interaction phenomena. A 496 simple interpretation of sorption mechanisms just based on reaction rate analysis is certainly 497 not feasible. This becomes obvious when looking to literature data. Exponential fitting of Tc 498 sorption data from Bondietti and Francis (1979) gives a k value of around 1.1±0.4 d⁻¹ for an initial Tc concentration of 1.1×10^{-7} M with Westerly granite as a solid material (167 g/L 499 500 solid/liquid ratio). The pH/Eh values for this material were also comparable (pH 8, Eh -100 501 mV) to the conditions used in present work, but neither Fe(II) content nor specific surface 502 area of the material were specified in the paper, which could significantly contribute to the 503 enhanced rates found in Bondietti and Francis (1979).

504 Distribution coefficients (K_d in L/kg) obtained for Tc sorption onto Äspö material were 505 calculated using the following equation (16):

$$K_{d} = \frac{C_l}{C_s} = \frac{A_0 - A_l}{A_l} \times \frac{V}{m_{solid}}$$
(16)

where C_l and C_s are the equilibrium concentrations of solutes in aqueous and solid phases, 507 respectively, A_0 and A_1 are the initial and final aqueous radionuclide activities at equilibrium 508 (Bq/mL), respectively, V is the volume of the aqueous phase (mL) and $m_{\it solid}$ is the solid 509 510 mass (g). Typical K_d values obtained within the present work are presented in Table 4 511 together with measured initial Tc concentrations, amount of ion-exchangeable Fe(II) and 512 redox potentials. From a thermodynamic point of view the K_d approach assumes a fully 513 reversible sorption process, but in most papers it is used even when irreversible 514 reduction/precipitation processes are involved (Albinsson et al., 1991; Allard et al., 1979; 515 Kaplan and Serne, 1998). In the report of USEPA (1999) authors describe "conditional" K_d 516 values for the interpretation of experimental data in cases when the rigorous application of 517 the K_d approach is prohibited (e.g. in non-equilibrium, irreversible, or solubility controlled systems). In the present work K_d values are considered as conditional distribution 518

519 coefficients. For a better comparison of the retention behaviour with literature data, K_d 520 should be normalized to the specific surface area of the solid material used in the 521 experiments according to the equation (17):

522

$$K_a = \frac{K_d}{S_{BET}} (\times 10^{-6}) \tag{17}$$

523 where K_a is the surface area normalized distribution coefficient (m) and S_{BET} is the specific surface area of the solid determined by N₂-BET (m^2/g). K_a values for the ÄD material are 524 525 shown in Table 4. Since the surface area of oxidized and un-oxidized AD remains the same, 526 this normalization does not affect the comparison between them. The K_d values obtained for the oxidized material (see Table 4; 10^{-5} M Tc) in this study with 0.53±0.05 L/kg for oxidized 527 ÄD are in good agreement with recently published data of Videnska and Havlova (2012) who 528 reported a K_d value of 0.3 L/kg under oxidizing conditions for 1.6×10⁻⁴M ⁹⁹Tc on granitic rocks 529 530 from Melechov Massive, Centre Bohemian Massive, Czech Republic (> 0.8 mm size fraction). 531 Since the surface area was not mentioned in the publication, K_a values cannot be calculated 532 for these results. Batch-type studies on Tc uptake with well-preserved un-oxidized crushed 533 granitic rocks can hardly be found in the literature. We have compared the data obtained 534 with experiments performed under artificial reducing conditions. Ito and Kanno (1988) have 535 published the Tc distribution between solution and granitic rocks (0.49 - 0.83 mm size fraction, 10⁻¹² M [^{95m}Tc]) together with a number of other minerals under oxidizing and 536 537 reducing conditions. Under oxidizing conditions (in 0.16 M NaNO₃) K_d values of 0.1 L/kg (K_a = 538 4.8×10^{-7} m) were obtained, while under reducing conditions (0.1 M NaBH₄ and 0.16 M NaNO₃) values of 45.6 L/kg ($K_a = 2.2 \times 10^{-4}$ m) increasing up to 68 L/kg ($K_a = 3.2 \times 10^{-4}$ m) with 539 540 decreasing NO_3^{-1} concentration down to 0.016 M could be determined. The paper by 541 McKinley and Scholtis (1993) summarized the data on K_d values of Tc on different materials 542 including granitic rocks used for the safety assessment of waste disposal at that time. Here 543 the values range from 0 to 250 L/kg depending on the experimental conditions and the 544 origin of the rocks. The authors emphasize that Tc sorption under oxidizing conditions is 545 normally very low or zero, and for reducing conditions K_d values increase by 1-2 orders of 546 magnitude.

547 **4.3.** Tc desorption

548 As was shown in the results section, Tc desorption without artificial oxidation of the material 549 is almost negligeable. We suppose that the relatively rapid establishment of a steady state 550 for the Tc concentration after only 1 day desorption is the consequence of washing out 551 residual Tc(VII) in the porewater rather than desorption of sorbed or precipitated Tc(IV). For 552 the desorption experiments after artificial oxidation, the general trend of the fast desorption 553 up to a plateau value might be explained by the rapid oxidation of surface associated Tc(IV) 554 to the heptavalent oxidation state and thus to the (almost) non-sorbing pertechnetate anion 555 whereas strongly sorbing un-oxidized Tc(IV) does not have any significant contribution to the 556 desorption processs. The same observations were made by numerious investigators (Begg et 557 al., 2008; Burke et al., 2006; Morris et al., 2008), where Tc(IV) oxidation to Tc(VII) was found 558 to be a driving force for the Tc desorption process. Taking into account Tc uptake 559 mechanisms discussed in section 4.1, one can assume two different mechanisms, namely 560 predominantly surface complexation for log $[Tc]_{tot}$ = -8 and -9 and $TcO_2 \cdot 1.6H_2O_{(s)}$ 561 precipitation for log [Tc]_{tot} = -5 governing the Tc sorption process. Consequently, for the 562 same initial Tc concentration, a similar Tc desorption behaviour can be expected for oxidized 563 and un-oxidized systems, which is in a good agreement with the experimental data observed 564 for the ÄD. A higher degree of desorption observed for the samples with higher [Tc]_{initial} 565 might be explained by the relatively easier oxidation of the surface associated TcO₂·1.6H₂O_(s) 566 phase in comparison to the surface complexed Tc(IV) species (see Table 5) which are 567 probably more recalcitrant to remobilization during oxidation. Similar presumptions were 568 expressed (Begg et al., 2008; Burke et al., 2006; Morris et al., 2008), but the mechanistical 569 nature of this resistance is still a matter of further research.

570	Table 5. Comparison of the sorption/desorption values for the oxidized and un-oxidized AD for the
571	artificially oxidized sample series. Desorption time of 1 month.

Material	-log [Tc] _{initial}	Tc sorption, %	Tc desorption, %
	5	21±2	93.7±0.6
ÄD (un-oxidized)	8	99.2±0.6	64±7
	9	99.5±6	33±7
ÄD (oxidized)	5	12±1	94±3

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8	47±8	64±8	
9	84±6	35±8	

572

573 **4.4.** Spectroscopic analyses

574 According to the XPS results (see Figure 5 left), the Tc signal was detected only on dark mica-575 type mineral surface (most likely biotite or magnetite), which contains structural Fe(II). It is 576 well known that Tc(VII) is concentrated on Fe(II)-containing minerals (Burke et al., 2010; 577 Fredrickson et al., 2009; McBeth et al., 2011) as "hotspots" due to surface (heterogeneous) 578 reduction into insoluble Tc(IV) oxide phase or/and homogeneous reduction in the solution 579 into soluble Tc(IV) species with subsequent sorption on the mineral surface. According to the 580 results of Peretyazhko et al. (2008a) heterogeneous Tc(VII) reduction on the surface 581 associated Fe(II) is orders of magnitude faster than the homogeneous reduction by aqueous 582 Fe²⁺ and may occur on the Fe(II)-containing mineral surfaces. XANES analysis of the Äspö 583 diorite contacted with Tc-containing ÄGWS indicated the presence of Tc(IV) as well.

584 **4.5.** Thermodynamic modelling

The general scheme of the modelling applied is described in Section 2.6. A similar calculation approach was applied in previous studies related to the modelling of Pu sorption to kaolinite (Marsac et al., 2015a) and Np sorption on illite (Marsac et al., 2015b) with respect to the differences in chemistry. Figure 7 shows the Tc(IV) fraction in the aqueous phase modelled with PHREEQC for log [Tc]_{tot} \leq -8 as a function of the redox potential. For pH = 8.0 and I = 0.1 M, Eh of the [Tc(VII)]_{aq}/[Tc(IV)]_{aq} borderline is -53 mV (Eh_{aq}, calculated from equation (5)), as highlighted in Figure 7. To account for sorption processes (see equation (9)), log K_{VII/IV,surf} is determined as follows. The site density is arbitrarily set equal to 1 site/nm², which is the same order of magnitude as commonly observed for several minerals (Burlakov et al., 2000; Jeppu and Clement, 2012). The value for log K_{VII/IV,surf} is strongly dependent on this arbitrarily chosen site density. At Eh values \leq -90 mV almost all Tc is calculated to be Tc(IV) (see Figure 8). Such conditions prevail in the experiments with un-oxidized ÄD and log [Tc]_{tot} \leq -8. Taking these experimental data, log K_{VII/IV,surf} is deduced by a fitting procedure and is found to be equal to 35.7 (see equation (8)). This value is kept constant for all the following calculations.



Figure 7. Tc(IV) fractions depending on the redox potential (pH 8, I = 0.1 M) calculated according to the equations (4) and (9), respectively. Eh_{exp} corresponds to the measured redox potential; Eh_{calc} relates to the calculated redox potential where 21% of Tc is removed from solution in the experiment with log $[Tc]_{tot} = -5$; $[Tc(VII)]_{aq}/[Tc(IV)]_{aq}$ and $[Tc(VII)]_{aq}/[Tc(IV)]_{surf}$ borderline redox potentials are as well marked by vertical dashed lines; the shaded area at the left side shows the Eh region where model calculations predict quantitative uptake.

585 The borderline redox potential where [Tc(IV)]_{surf} and [Tc(VII)]_{aq} exist in equimolar 586 concentrations, Eh_{surf}, is found equal to -12 mV in accordance to the equation (10). In 587 agreement with the findings of Marsac et al. (Marsac et al., 2015a; Marsac et al., 2015b) for 588 Np and Pu sorption to clays, the strong sorption behaviour of Tc(IV) shifts the borderline Eh 589 to higher values as compared to that for the $[Tc(IV)]_{aq}/[Tc(VII)]_{aq}$ ratio, Eh_{aq} , in aqueous 590 solution in the absence of mineral surfaces. For log $[Tc]_{tot} = -5$ precipitation of $TcO_2 \cdot 1.6H_2O_{(s)}$ is predicted, which represents the 2nd retention mechanism, and extends the stability field of 591 Tc(IV) further. In the respective experiment the measured Eh value (Eh_{exp}) is 76 mV ± 50mV, 592

593 where the model predicts insignificant uptake in disagreement with the experimentally 594 observed $\approx 21\%$ uptake. Actually, the model is very sensitive to the redox potential. For 595 instance, a change in Tc uptake from 2% to 80% is predicted by assuming a slight decrease in 596 Eh_{calc} from +20 to 0 mV for log $[Tc]_{tot}$ = -5. Regarding the uncertainties of experimental Eh 597 data in a range of ±50 mV (Altmaier et al., 2011a) and the incertitude of model parameters, 598 our simulation results are not so different from experimental data. The discrepancy of Tc 599 uptake in simulation and experiment can easily be explained by uncertainties in redox 600 potential measurements. According to model calculations, the dominating uptake 601 mechanism at log $[Tc]_{tot}$ = -5 is precipitation of $TcO_2 \cdot 1.6H_2O_{(s)}$. This is in agreement with 602 XANES measurements performed with a 1mM Tc-containing solution contacted with the 603 crushed material. The model predicts almost 100% Tc uptake for Eh \leq -90 mV in accordance 604 with the experimental results for un-oxidized ÄD at log [Tc]_{tot} ≤ -8 (see Table 4). However, 605 relatively high Tc uptake (> 30%, see Table 4 & 5) of the oxidized materials at log $[Tc]_{tot} \leq -8$ 606 and rather high Eh values (> 200 mV) cannot be explained by the thermodynamic model. No 607 Tc reduction and no uptake is calculated above $Eh \ge \approx 30 \text{ mV}$ (see Figure 7). Nevertheless, Tc 608 uptake by the oxidized materials is significant but is lower as compared to that by the un-609 oxidized ÄD (see Figure 2). This finding may suggest that, although oxidized, a redox partner 610 for the reduction of Tc(VII) to Tc(IV) is present in significant amounts, which, however, is not 611 detected by the Eh measurements using a Pt-electrode. Another aspect relates to the concentration of redox partners required for reproducible and thermodynamically defined 612 redox potential measurements using a Pt-electrode, which is proposed to be $\geq \approx 10^{-6}$ M 613 614 (Grenthe et al., 1992). The ion-exchangeable ferrous iron concentration in the oxidized rock materials lie very close to that minimum value of (9±5)×10⁻⁶ M Fe(II) for oxidized ÄD. It 615 616 seems likely that Fe(II) content available is quantitatively consumed during the Tc reduction 617 process by oxidization to Fe(III) inhibiting a precise Eh determination by the platinum 618 electrode. Thus a higher uncertainty in the Eh measurements and in the interpretation of the 619 results needs to be considered for the higher [Tc]_{tot} series and the oxidized ÄD.

620

621 **5. Conclusions**

623 A batch type sorption/desorption study on Tc interaction with crushed granitic rock material 624 from the generic URL in Sweden (Äspö HRL) was performed. Part of the crushed material 625 from Äspö HRL used in the experiments has been obtained and handled under anoxic 626 conditions before and during the experiments thus representing to a great extent natural in-627 situ rock material conditions (e.g. redox capacity) relevant to the far-field environment of a 628 repository in crystalline rock. According to the experimental findings, Tc(VII) is reduced 629 under natural anoxic conditions to Tc(IV) followed by either precipitation of TcO₂·1.6H₂O_(s) 630 or/and surface complexation of soluble Tc(IV) species on the rock surface. Comparison of 631 samples after ultracentrifugation shows no detectable amount of Tc colloidal phase under 632 simulated groundwater conditions applied. After contact of Tc(VII)-containing groundwater 633 simulants with crystalline rock materials, Tc(IV) species were detected on mafic mica-type 634 Fe-containing minerals. Apparently, the Tc(VII) concentration directly influences the amount 635 of Tc uptake on un-oxidized and oxidized material, which can be correlated with the ion-636 exchangeable Fe(II) buffer available $(4 - 6 \mu g/g \text{ for un-oxidized } AD and 1 - 3 \mu g/g \text{ for } and 1 - 3 \mu g/g \text{ for } and 1 - 3 \mu g/g \text{ for } and a negative for a$ oxidized ÄD). Tc behaviour on oxidized ÄD differs dramatically compared to un-oxidized ÄD 637 638 samples. In general, artificially oxidized rock material could retain approx. 2 times less Tc 639 compared to un-oxidized material. Distribution coefficients (K_d) and surface area normalized 640 distribution coefficients (K_a) were determined for artificially oxidized and original un-oxidized 641 ÄD which serve as parameters for transport modelling or performance assessment modelling 642 purposes. Technetium desorption from the ÄD is insignificant under anoxic conditions, but 643 after artificial oxidation technetium mobility is increased due to re-oxidation of Tc(IV) to 644 Tc(VII). Such reactions might become relevant in scenarios where oxidized glacial melt water 645 intrusion into a repository is considered. The similar Tc desorption behaviour from the 646 initially oxidized and un-oxidized ÄD samples indicates the same retention mechanism for 647 both types of the material even taking into account the difference in the total uptake 648 capacity. Samples from the batch sorption studies with high [Tc]_{tot} show increased 649 desorption most probably due to faster re-oxidation of the TcO₂·1.6H₂O_(s) precipitate in 650 comparison to surface complexed Tc(IV) species. According to the thermodynamic 651 calculations, for log $[Tc] \leq -8$ Tc(IV) surface complexation is a predominant process of Tc 652 immobilization, while at higher Tc concentration TcO₂·1.6H₂O_(s) precipitation plays the main

653 role. This work clearly highlights the importance of using well-preserved (un-oxidized) 654 natural materials for the reliable estimation of the interaction of redox sensitive elements 655 with these solid phases. The difference in retention capacity between artificially oxidized and 656 un-oxidized material is highly significant. The data obtained gives important implications on 657 the prediction of Tc behaviour under natural conditions for safety assessment of deep 658 geological disposals of SNF and HLW in undisturbed and disturbed crystalline environments.

659 6. Acknowledgements

660

661 The research leading to these results has received funding from the Federal Ministry of 662 Economics and Technology (BMWi) collaborative project VESPA (Behavior of long-lived fission and activation products in the near field of a repository and possible retention 663 664 mechanisms) under contract № 02 E 10800 and the European Union's European Atomic Energy Community's (Euratom) 7th Framework Programme FP7/2007-2011 under grant 665 666 agreement № 269658 (CP CROCK).

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669 **7. References**

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Supplementary Material



Figure A.1. (Top and middle) Cores used in the batch experiments directly after drilling. (Bottom) Two-layers packing of the cores, sealed into LD-PE and Al bags.

Table A.1. XRF data on Äspö diorite (material used in this study, old oxidized ÄD samples
849 used in (Huber et al., 2010 & 2012), oxidized ÄD from (Byegård et al., 1998)), and typical
850 granodiorite (Nockolds, 1954) composition.

granoulonic	(10000003, 1554) 0011	iposition.		
Element	Äspö diorite (this	Äspö diorite,	Äspö diorite	Typical
(RMS _{rel})	study), wt. %	(Huber et al.,	(Byegård et al.,	granodiorite
		2010 & 2012),	1998) <i>,</i> wt. %	(Nockolds,
		wt. %		1954), wt. %
SiO ₂	62.71	66.06	60.1	66.88
(0.4%) Al ₂ O ₃	17.27	16.89	18.1	15.66
(0.7%) Fe ₂ O ₃	4.39	2.6	5.1	1.33
(0.8%) FeO	2.51	0.87	- Q_	2.59
(0.8%) MnO	0.08	0.05	0.1	0.07
(1.8%) MgO	1.76	0.8	2.2	1.57
(0.3%) CaO	3.75	2.41	4.4	3.56
(0.5%) Na ₂ O	4.55	4.91	4.7	3.84
(1.3%) K ₂ O	3.05	4.38	3.2	3.07
(0.8%) TiO ₂	0.66	0.35	0.8	0.57
(1.0%) P ₂ O ₅	0.24	0.12	0.3	0.21
(1.1%) Loss on ignition	0.67	1.37	1.0	0.65
(LOI) Sum	99.1	99.4	100	100
Trace	Concentration,	Concentration,	Concentration,	
element (LLD)	ppm	ppm	ppm	
Ba (10.5ppm)	1162	n.d.	1770	
Со	11	5	n.d.	
(2.2ppm) Cr	24	6	n.d.	
(5.3 ppm) Cu	2	1	n.d.	
(1.6 ppm) Ga	23	19	n.d.	
(1 ppm) Nb	15	9	n.d.	
(1.1 ppm) Ni	18	6	n.d.	
(3.7 ppm) Pb	17	16	89	
(2.7 ppm)				

	ACCH	EPTED MANUSC	RIPT	
Sc	6	4	n.d.	
(2.7 ppm) Sr	1052	770	1300	
(1.5 ppm) Th	9.5	5.7	5.2	
(1.7 ppm) U	4.4	1.8	2.1	
(1.1 ppm) V	62	36	n.d.	<u>_</u>
(6.1 ppm) Y	22	16	18.1	
(1.3 ppm) Zn	76	43	n.d.	2
(1.1ppm) Zr	168	139	256	

n.d. – not detected, RMS_{rel} – relative root square mean, LLD – lower limit of detection; $\mathrm{RMS}_{\mathrm{rel}}\,\mathrm{and}\,\mathrm{LLD}$ are taken from the ÄD XRF analysis.

Mate-	Con-	Tc uptak	ke. %		_		_		_		_				
riai	tact	log [Tc]	= -5	log [Tc]	= -6	log [Tc]	= -7	log [Tc]	= -8	log [Tc]	= -9	log [Tc]	= -10	log [Tc]	= -11
	days	%	±. %	%	±. %	%	±. %	%	±. %	%	±. %	%	±. %	%	±. %
un-oxi	- 0.04	7.10	0.06					3.57	2.16	14.51	8.08				
dized	1	8.38	0.06					13.42	2.00	25.14	7.59				
AD	1	12.80	0.06					10.82	2.02	27.94	7.59				
	4	19.94	0.06					19.04	1.95	33.15	7.40				
	4	6.47	0.06					23.95	1.89	27.54	7.54				
	7	35.15	0.05					41.64	1.67	35.55	7.33				
	7	11.54	0.06					29.96	1.82	25.95	7.59				
	15	15.37	0.08					76.46	1.15	52.74	6.85				
	15	33.37	0.07					62.74	1.38	53.14	6.83				
	21	25.12	0.08					83.88	1.00	53.14	6.83				
	21	19.14	0.08					73.74	1.20	86.73	5.76				
	30	22.43	0.08					93.48	0.77	71.13	6.28				
	57	13.90	0.09					79.66	1.09	70.35	6.31				
	57	39.59	0.07					95.86	0.70	97.14	5.41				
	91	20.26	0.08					97.56	0.65	96.34	5.43				
	91	22.63	0.08					97.90	0.64	94.34	4.98				
	141	44.76	0.12	48.19	0.19	48.70	0.24	53.83	0.23	90.85	0.51	81.65	1.73	95.06	4.15
	141	46.67	0.12	25.49	0.16	41.41	0.21	85.39	0.40	90.34	0.49	76.36	0.99	95.65	4.66
	178	20.70	0.08					99.44	0.58	100.00	5.12				
	178							98.88	0.60	99.54	5.12				
oxi-	0.04	5.69	0.09				7	7.61	2.06	-3.45	8.35				
dized	0.04							4.02	2.10	8.15	8.06				
AD	1	9.40	0.09					7.65	2.06	18.96	7.78				
	1	9.31	0.09					8.59	2.05	-5.84	8.39				
	4	7.08	0.09		/			12.12	2.02	9.76	8.01				
	4	9.45	0.09					11.74	2.02	10.56	7.99				
	7	8.83	0.09					23.04	1.90	5.35	8.11				
	7	9.69	0.09					15.23	1.98	14.95	7.87				
	15	6.75	0.09					14.51	1.99	25.36	7.61				
	15	8.59	0.09					16.21	1.97	24.95	7.61				
	21	13.09	0.09					2.57	2.11	33.35	7.40				
	21	10.08	0.09					13.49	2.00	13.36	7.92				
	30	10.38	0.09					11.23	2.03	22.96	7.68				
	57	9.90	0.09					76.91	1.14	39.35	7.23				
	57	9.82	0.09					43.11	1.66	42.55	7.14				
	91	10.51	0.09					53.55	1.52	23.37	7.66				
	91	12.21	0.09					36.51	1.74	57.77	6.71				
	141	3.07	0.71	13.79	0.15	17.12	0.17	18.52	0.17	77.49	2.29	80.46	2.51	67.14	1.23
	141	1.64	0.09	8.56	0.15	16.69	0.17	16.96	0.17	24.18	1.11	63.70	1.17	60.10	1.02

858 **Table A.2.** List of samples from the batch sorption studies and long-term sorption 859 experiments on un-oxidized and oxidized ÄD.

	Mate-	Con-	Tc uptal	ke. %												
	rial	tact	log [Tc]	= -5	log [Tc]	= -6	log [Tc]	= -7	log [Tc]	= -8	log [Tc]	= -9	log [Tc]	= -10 log	g [Tc]	= -11
		time. days	%	±. %	%	±. %	s %	±. %	s %	±. %	%	±. %	%	±. %	%	±. %
		178							54.83	1.82	90.15	5.67				
		178	11.69	0.09					39.70	1.82	77.75	6.07				
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Figure A.2. ÄD sample for XPS after contacting with Tc(VII)-containing ÄGWS. Red circle indicates the region where Tc(IV) was found.

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Paper highlights:

- Tc sorption/desorption kinetics investigation was performed on both oxidized and unoxidized Äspö diorite (Sweden) on different Tc concentrations.
- The difference between artificially oxidized and well-preserved natural rock material is shown.
- Tc colloidal phase formation was not detected under simulated natural conditions.
- Spectroscopic investigations shown the accumulation of Tc(IV) species on mafic (e.g. mica, magnetite) minerals
- Tc sorption parameters on the rock material studied (distribution coefficient, sorption rate) were derived.

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