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H-bonding scheme and cation partitioning in axinite: a single-crystal neutron diffraction and Mössbauer spectroscopic study

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Abstract The crystal chemistry of a ferroaxinite from Colebrook Hill, Rosebery district, Tasmania, Australia, was investigated by electron microprobe analysis in wavelength-dispersive mode, inductively coupled plasma–atomic emission spectroscopy (ICP–AES), ⁵⁷Fe Mössbauer spectroscopy and single-crystal neutron diffraction at 293 K. The chemical formula obtained on the basis of the ICP–AES data is the following: X1,X2Ca_{4.03} Y (Mn_{0.42}Mg_{0.23}Fe²⁺_{1.39}) $_{\Sigma2.04}$ Z1,Z2 (Fe³⁺_{0.15}Al_{3.55}Ti_{0.12}) $_{\Sigma3.82}$ T1,T2,T3,T4 (Ti_{0.03}Si_{7.97}) $_{\Sigma8}$ T5

 $B_{1.96}O_{30}(OH)_{2.18}$. The ⁵⁷Fe Mössbauer spectrum shows unambiguously the occurrence of Fe²⁺ and Fe³⁺ in octahedral coordination only, with Fe²⁺/Fe³⁺ = 9:1. The neutron structure refinement provides a structure model in general agreement with the previous experimental findings: the tetrahedral *T1*, *T2*, *T3* and *T4* sites are fully occupied

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by Si, whereas the T5 site is fully occupied by B, with no evidence of Si at the T5, or Al or Fe^{3+} at the T1–T5 sites. The structural and chemical data of this study suggest that the amount of B in ferroaxinite is that expected from the ideal stoichiometry: 2 a.p.f.u. (for 32 O). The atomic distribution among the X1, X2, Y, Z1 and Z2 sites obtained by neutron structure refinement is in good agreement with that based on the ICP-AES data. For the first time, an unambiguous localization of the H site is obtained, which forms a hydroxyl group with the oxygen atom at the O16 site as donor. The H-bonding scheme in axinite structure is now fully described: the O16-H distance (corrected for riding motion effect) is 0.991(1) Å and an asymmetric bifurcated bonding configuration occurs, with O5 and O13 as acceptors [i.e. with O16...O5 = 3.096(1) Å, H...O5 = 2.450(1) Å and $O16-H.O5 = 123.9(1)^{\circ}$; O16.O13 = 2.777(1) Å, $H \cdots O13 = 1.914(1)$ Å and $O16 - H \cdots O13 = 146.9(1)^{\circ}$].

Keywords Axinite · Single-crystal neutron diffraction · Mössbauer spectroscopy · H-bonding scheme · Cation partitioning

Introduction

The axinite-group minerals are chemically (and structurally) complex borosilicates with general formula $(Ca,Mn)_4$ $((Mn,Fe,Mg)_2)_2Al_4B_2Si_8O_{30}(OH)_2$. Axinites typically occur in low- to medium-grade metamorphic environments, in particular in Ca- and B-rich (but relatively Al-poor) environments, and the different $P-T-X-f_{O2}$ conditions govern the remarkable compositional variability of this mineral group (Pringle and Kawachi 1980; Grew 1996).

Axinite often occurs in large crystals (mm- to cm-sized), which have a relatively good hardness (ranging from 6.5 to

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7 on the Mohs scale). One of the main uses of axinite is in the jewellery market, because it is one of the few gemstones that possess a strong pleochroism and high brilliance (i.e. cut crystals are usually intensely trichroic, with the brown and purple colours dominating; birefringence 0.010–0.012), well-appreciated features for a jewel. Crystals of axinite are transparent to translucent and have a distinctively strong vitreous lustre when cut and polished. Cut and clear axinite larger than 5 carats is very rare. The ability to fix up the provenance of a gem-quality axinite stock, through a detailed crystal-chemistry characterization of some of its samples, is a key point to provide a correct economic assessment of such a material on the gemstones market.

The first structure model of axinite was reported by Ito and Takéuchi (1952) and later revised by Ito et al. (1969) and Takéuchi et al. (1974) in the space group P-1 with $a \sim 7.16$ Å, $b \sim 9.20$ Å, $c \sim 8.97$ Å, $\alpha \sim 91.8^{\circ}$, $\beta \sim 98.1^{\circ}$ and $\gamma \sim 77.2^{\circ}$. Structural investigations of different members of the axinite group were successively reported by several authors (e.g. Swinnea et al. 1981; Belokoneva et al. 1997, 2001; Andreozzi et al. 2000a, b, 2004; Filip et al. 2006). The topology of the axinite structure is complex. The structure is built up by a sequence of alternating sheets of tetrahedra and octahedra, as shown in Fig. 1. In the tetrahedral sheets, there are two different building units: (a) two disilicate groups are connected by two B-tetrahedra to form a six-membered ring and (b) two further disilicate groups share a corner with the B-tetrahedra forming a complex $[B_2Si_8O_{30}]$ unit. In the octahedral sheets, edge-sharing distorted octahedra, populated by Al, Mn, Fe or Mg, form six-membered finite chains, which are laterally connected by distorted Ca-octahedra.

A few studies were devoted to the classification of axinite-group minerals (Sanero and Gottardi 1968) and to their physical properties (e.g. Lumpkin and Ribbe 1979; Filip et al. 2008). Andreozzi et al. (2004) proposed a new crystal-chemical formula for the axinite mineral group:

$$^{VI}[X1X2YZ1Z2]_2^{IV}[T1T2T3T4T5]_2O_{30}(OwOH_{1-w})_2$$

where VI and IV are coordination numbers; XI = Ca and very minor Na; X2 = Ca (in axinites) or Mn (in tinzenite); Y = Mn (in manganaxinite and tinzenite), Fe²⁺ (in ferroaxinite) or Mg (in magnesioaxinite), with minor Al and Fe³⁺; ZI = Al and Fe³⁺; Z2 = Al; TI, T2 and T3 = Si; T4 = Si(and presumably very minor B); T5 = B and minor Si. Charge unbalance (w), due to heterovalent substitutions, is compensated for by O²⁻ \rightarrow OH⁻ substitution.

The structural refinements of different members of the axinite group were all based on X-ray diffraction data (e.g. Takéuchi et al. 1974; Swinnea et al. 1981; Belokoneva et al. 1997, 2001; Andreozzi et al. 2000a, b, 2004; Filip et al. 2006). On the basis of literature data, a poorly defined picture of the configuration of the O–H groups and of the H-bonding scheme still exists. Andreozzi et al. (2000a)

studied the crystal chemistry of 60 samples of axinites from 24 worldwide localities by single-crystal X-ray diffraction, electron and ion microprobe, and ⁵⁷Fe Mössbauer spectroscopy. These authors showed evidence that B (instead of Al) is inversely related with Si, and that Fe³⁺ is the main substitute for Al, but also for divalent cations, and that these heterovalent substitutions are balanced by an OH deficiency. Moreover, Andreozzi et al. (2004) highlighted how Fe partitioning among octahedral and/or tetrahedral sites must be defined for a reliable description of the crystal chemistry of axinites. Iron in axinite is reported to occur both in the divalent and in the trivalent valence state. Quantification of the Fe³⁺/Fe²⁺ ratio is reliably done using ⁵⁷Fe Mössbauer spectroscopy, which also allows one, on the basis of the hyperfine parameters, to discern between octahedral and tetrahedral coordination. This method was applied several times to study axinite, among others by Pieczka and Kraczka (1994), Fuchs et al. (1997), Andreozzi et al. (2000a, b, 2004), Novák and Filip (2002), Zabinski et al. (2002) and Filip et al. (2006). Most of these studies show that Fe^{2+} is dominating in ferro- and magnesioaxinite, and, in this condition, one single doublet is sufficient to model the Fe^{2+} resonance absorption contribution. In contrast, Fe³⁺ is the dominant form of iron in manganaxinite and tinzenite (Filip et al. 2008). Following Andreozzi et al. (2004), Fe²⁺ populates the Y site. For Fe^{3+} some contradicting interpretations are found. Fuchs et al. (1997) attributed it to the tetrahedral site (even if isomer shift values would be unduly large for Fe^{3+} in tetrahedral coordination), whereas, according to a more shared opinion, Fe³⁺ is supposed to enter the octahedral sites (especially the Z1 site and to a lesser extent the Y site; Andreozzi et al. 2004; Filip et al. 2006).

The present study re-investigates the crystal structure and crystal chemistry of an axinite from Colebrook Hill, Rosebery district, Tasmania, Australia, with the aim to provide a reliable location of the proton site for a full description of the atomic relationship via the H-bond. To achieve this goal and obtain a clear picture of cations ordering at the tetrahedral and octahedral sites, we used electron microprobe analysis in wavelength-dispersive mode (EPMA-WDS), inductively coupled plasma–atomic emission spectroscopy (ICP–AES), ⁵⁷Fe Mössbauer spectroscopy and single-crystal neutron diffraction (SCND). This investigation is the latest in a series of single-crystal neutron diffraction experiments that we have recently devoted to hydrous B-bearing minerals (e.g. Gatta et al. 2010, 2012a, b, 2014a).

Sample description and mineralogy

We have selected a sample of axinite from Colebrook Hill, Rosebery district, Tasmania, Australia, provided by the Museum of Mineralogy of the University of Padova





Fig. 1 Crystal structure of ferroaxinite viewed down [100], [010] and [001], based on the neutron structure refinement of this study. An additional clinographic view shows the sequence of alternating sheets

(Italy), ref. MMP M13333. The sample utilized for this study contains large laminar, brown crystals of axinite up to several millimetres in length, lined in a cavity associated with quartz crystals, dissolved from the enclosing calcite with acids. The rock matrix is composed by an aggregate of microcrystalline axinite and green diopside. Back-scattered images obtained on a polished section of axinite-(Fe) (collected by a scanning electron microprobe) have shown diffused micrometric inclusions of titanite as well.

of tetrahedra and octahedra, which form the structure of axinite. Atomic displacement probability factor: 50 %

Colebrook Hill mine has produced Australia's best axinite specimens (ferroaxinite), some of which are world class. The deposit is hosted by an unusual skarn, sometimes described as a limurite or axinite-hornfels, probably derived from calcareous or dolomitic rocks associated with mafic–ultramafic intrusive rocks, and these latter formed by sills and dykes of serpentinized pyroxenite and gabbro (Blisset 1962). The skarn is supposed to derive from boron and sulphur-rich reactive Table 1Representativecomposition of the ferroaxinitefrom Colebrook Hill, based onEPMA-WDS analysis (averageof 20 data points)

	wt%	e.s.d.		a.p.f.u.
MgO	0.33	0.08	Mg	0.09
CaO	19.10	0.3	Ca	3.86
MnO	2.43	0.1	Mn^{2+}	0.39
FeO	9.16	0.2	Fe ²⁺	1.45
Fe ₂ O ₃	0.89	0.2	Fe ³⁺	0.14
B ₂ O ₃	6.11	0.15	B^{3+}	2.00
Al_2O_3	18.15	0.2	Al	4.04
SiO ₂	42.06	0.8	Si	7.94
TiO ₂	0.04	0.02	Ti^{4+}	0.01
H ₂ O ^a	1.73		OH-	2.18
Total	100.00			

The chemical formula (calculated on the basis of 32 oxygen atoms and with Fe²⁺ /Fe³⁺ = 9:1, as suggested by the Mössbauer data) is the following: $x_{1,X2}(Ca_{3.86}Mn_{0.07})_{\Sigma 3.93}$ $^{Y}(Mn_{0.32}Mg_{0.09}Fe_{1.45}^{2+}Fe_{0.14}^{3+})_{\Sigma 2.00}$

 Z1,Z2 Al_{4.00} T1,T2,T3,T4 (Al_{0.04}Ti_{0.01}Si_{7.94}) $_{\Sigma7.99}$ T5 B₂O₃₀(OH)_{2.18}

^a Calculated by difference to 100 wt%; B₂O₃ calculated from the ideal stoichiometry

hydrothermal solutions from an underlying granite of Devonian age.

Colebrook Hill mine was originally quarried for copper; indeed, the most abundant mineral is ferroaxinite. The main metallic minerals are pyrrhotite, pyrite, chalcopyrite and arsenopyrite. Small amounts of marcasite, galena, sphalerite and tetrahedrite were also observed. Additional boron-bearing silicates, represented by datolite and danburite, occur in vugs as well, but are rare. Gangue minerals are common and formed by calcite and quartz. Banded ore consists of alternating bands of axinite, actinolite and diopside. Much of the sulphides, calcite and quartz are associated with the axinite, with crystals up to 2-3 cm in length lined in the cavities filled with calcite. The sulphides usually occur as irregular patches throughout the axinite veins or in bands between the axinite and the actinolite veins. The massive ore is made of crystals of ferroaxinite, actinolite, diopside, calcite and quartz. The sulphides, in particular pyrrhotite and chalcopyrite, occur either disseminated through the ore or as irregular veins (Petterd 1910; Blissett 1962).

Experimental methods

Chemical analyses

Scanning electron microscope observations were performed using a SEM JEOL-5610 LV at the Museum of Natural History of Milan, Italy. Chemical composition of ferroaxinite was obtained by a CAMECA SX-50 electron microprobe equipped with four wavelength-dispersive spectrometers and one energy-dispersive spectrometer at the laboratory of microanalysis of the Institute for Geosciences and Earth Resources of CNR (Padova, Italy). The operating conditions were 20 kV accelerating voltage and 20 nA beam current. Counting times were 10 s at the peak and 5 s at the background for major elements Mg, Al, Si, Ca, Mn and Fe. Emission X-ray counts were converted into oxide weight percentages using the PAP correction program (Pouchou and Pichoir 1985). Crystals were found to be chemically homogeneous within the analytical error. The chemical formula, obtained by averaging 20-point analyses, is given in Table 1.

A total mass of 762 mg of the axinite sample was used for the ICP–AES chemical analyses. Sample preparation, standardization, quantification of uncertainty in measurement and analytical methods were performed using the guidelines of the following international protocols: EN ISO 21587 (2007) (parts 1, 2 and 3), 26845 (2008), 21078 part 1 (2008); EPA 3052 (1996), 6010c (2007), 9214 (1996); JGCM 100 (2008), as reported by Gatta et al. (2014b). Sample preparation was carried out in two ways:

- 1. Microwave-assisted acid attack at 200 °C with a mixture of sulphuric acid, hydrochloric acid and hydrofluoric acid; subsequent complexation of fluorides with boric acid;
- Decomposition by alkaline fusion with sodium carbonate, potassium carbonate or mixture of sodium carbonate and boric acid, in a muffle furnace at 1000–1100 °C for 15 min. Subsequent removal of silica by means of hydrofluoric acid and dissolution in mixture of hydrochloric acid and sulphuric acid.

Table 2Chemical compositionof the ferroaxinite sample basedon ICP-AES analysis

Oxides	wt%	e.s.d.	λ ICP/AES (nm)	Alkaline fusion—ICP/AES	Microwave-assisted acid digestion—ICP/AES	Wet method
SiO ₂	41.9	0.7	251.611	х		(1)
FeO	9.71	0.3	238.204	х	Х	
MnO	2.62	0.15	257.610	Х	Х	
Cr ₂ O ₃	0.09	n.d.	267.716	х	Х	
TiO ₂	1.07	0.25	334.940	х	Х	
Al_2O_3	16.0	0.5	396.153	Х	Х	(2)
CaO	19.8	0.6	317.933	х	Х	
MgO	0.81	0.20	285.213	х	Х	
B_2O_3	5.97	0.30	208.957	Х		
			249.677	Х		
K ₂ O	0.03	n.d.	766.490	Х	Х	
ZnO	0.06	n.d.	206.200	Х	Х	
SrO	0.06	n.d.	407.771	х	Х	
P_2O_5	0.05	n.d.	213.617	х	Х	
H_2O^a	1.83					
Total	100.00					

The chemical formula (calculated on the basis of 32 oxygen atoms and with Fe²⁺ / Fe³⁺ = 9:1, as suggested by the Mössbauer data) is the following: ${}^{X1,X2}Ca_{4.03}$ ${}^{Y}\left(Mn_{0.42}Mg_{0.23}Fe^{2+}_{1.39}\right)_{\Sigma 2.04}$ ${}^{Z1,Z2}\left(Fe^{3+}_{0.15}Al_{3.55}Ti_{0.12}\right)_{\Sigma 3.82}$ ${}^{T1,T2,T3,T4}(Ti_{0.03}Si_{7.97})_{\Sigma 8.00}$ ${}^{T5}B_{1.96}O_{30}(OH)_{2.18}$ (1) gravimetric method; (2) CyDTA-Zinc back titration method

^a By difference to 100 wt %; Li₂O, BeO and Na₂O contents were below the detection limit

Major elements, as Si and Al, were determined by at least two different methods: Si content was determined by gravimetrical method and ICP-AES; Al content was determined by ICP-AES and volumetric method, with separation by cupferron with addition of CyDTA in excess and retro-titration with zinc standard solution. B was measured by means of ICP-AES. Ca, Fe, Mn and other minor elements were determined by ICP-AES on aqueous solutions resulting from acid attack and alkaline fusion. Several analytical standard solutions were used to make a calibration curve, and various certified material standards were used as analytical control. Weight determination was carried out by means of an Analytical Balance GENIUS-Sartorius ME235P. A Berghof speedwave 4 apparatus was used as microwave digestion system. ICP-AES chemical analyses were performed by a Perkin Elmer Optima 7000 DV spectrometer. In this apparatus, the sample-introduction unit includes a cyclonic spray chamber and a Meinhard-type concentric glass nebulizer. The cyclonic spray chamber was used to provide both high sample transfer into the ICP and fast sample rinsein and rinse-out times. The concentric nebulizer provided excellent sensitivity and precision for aqueous solutions. The final chemical composition of the axinite, along with further details pertaining to the analytical methods, is given in Table 2.

Single-crystal neutron diffraction experiment

Single crystals of axinite, optically free of inclusions or fractures under a transmitted-light polarizing microscope, were selected for the diffraction experiments. The unit-cell parameters at room temperature were first measured by single-crystal X-ray diffraction (using a small crystal 0.28 × 0.18 × 0.12 mm) with an Oxford Diffraction four-circle diffractometer equipped with a CCD, at the Earth Sciences Dept, University of Milan (Italy). A total of 1980 reflections were measured, giving a metrically triclinic unit-cell with: a = 7.1629(2) Å, b = 9.2019(3) Å, c = 8.9608(3) Å, $\alpha = 91.792(2)^\circ$, $\beta = 96.66(4)^\circ$, $\gamma = 77.307(3)^\circ$ and V = 570.26(3) Å³ (Table 3).

A suite of crystals was sorted out for the neutron diffraction experiment. The diffraction patterns showed that mosaicity affects all the crystals of the axinite sample used in this study. A monochromatic single-crystal neutron diffraction experiment was performed on a crystal of axinite $(3.2 \times 3.1 \times 2.5 \text{ mm})$ using the hot source (fast neutrons) diffractometer HEiDi at the neutron source FRM II of the Heinz Maier-Leibnitz-Zentrum (MLZ), Germany. A first set of diffraction data were collected at 293 K, with a wavelength of the incident beam of 1.170(1) Å up to $2\theta_{\text{max}} = 124^{\circ} [\sin(\theta)/\lambda = 0.75 \text{ Å}^{-1}]$ [Ge-311 monochromator, Er foil filter to suppress $\lambda/3$ contamination]. A ³He
 Table 3 Data pertaining to the neutron data collection and structure refinements of ferroaxinite

Т(К)	293
Crystal size (mm)	$3.2 \times 3.1 \times 2.5$
Neutron radiation	Constant wavelengths:
	A: $\lambda = 1.170(2)$ Å, B: $\lambda = 0.7925(5)$ Å
Reference formula	$Ca_4(Mn,Fe)_2Al_4B_2Si_8O_{30}(OH)_2$
Ζ	1
Scan details	
λ , time/step, collimation, range, type, steps (u , v , q)	A: 1–6 s, 60', $6.5^{\circ} < 2\theta < 80^{\circ}, \omega$ -scan, 41 steps (17.4, –66, 79) $80^{\circ} < 2\theta < 100^{\circ}, \omega$ - θ -scan, 31 steps (18, –65, 85) $100^{\circ} < 2\theta < 124^{\circ}, \omega$ - 2θ scan, 25 steps (15, 0, 0) B: 2–8 s, 60', $75^{\circ} < 2\theta < 90^{\circ}, \omega$ - θ scan, 35 steps (12.25, 0, 0)
Space group	<i>P</i> -1
<i>a</i> (Å)	7.1629(2)
b (Å)	9.2019(3)
<i>c</i> (Å)	8.9608(3)
α (°)	91.792(2)
β (°)	98.227(3)
γ (°)	77.307(3)
d_{\min} (Å)	0.56
$V(Å^3)$	570.26(3)
$h_{\min}/h_{\max}, k_{\min}/k_{\max}, l_{\min}/l_{\max}$	-12/+12, $-16/+16$, $-16/+16$
$d_{\min}(\text{\AA})$	0.56
Measured reflections	6525 (5724 + 801)
Reflections with $F_{\rm O} > 4\sigma(F_{\rm O})$	5440
$R_{ m Friedel}, R_{ m \sigma}$	0.0271, 0.0445
No. of refined parameters	251
R_1 , observed reflections	0.0320
R_1 , all reflections	0.0486
wR^2 , all reflections	0.0611
GoF	1.398
Extinction coefficient	0.0189(9)
Residuals (fm/Å ³)	-0.86/+0.87

 $R_{\text{int}} = \Sigma \left| F_{\text{obs}}^2 - F_{\text{obs(mean)}}^2 \right| / \Sigma \left[F_{\text{obs}}^2 \right], \text{ based on the Friedel pairs; } R_1 = \Sigma (|F_{\text{obs}}| - |F_{\text{calc}}|) / \Sigma |F_{\text{obs}}|; \\ wR_2 = \left[\Sigma \left[w (F_{\text{obs}}^2 - F_{\text{calc}}^2)^2 \right] / \Sigma \left[w (F_{\text{obs}}^2)^2 \right]^{0.5}, w = 1 / \sigma^2 (F_{\text{obs}}^2) + (0.01 * P)^2 \right], P = (\text{Max}(F_{\text{obs}}^2, 0) + 2 * F_{\text{calc}}^2) / 3 \\ \text{Unit-cell parameters based on single-crystal X-ray data (1980 reflections, CCD). Isotropic extinction coefficient based on the Larson's formalism (Larson 1967), as implemented in SHELX-97 \\ \text{Scan width} = (u + v \times \tan\theta + q \times \tan^2 \theta)^{0.5}$

single counter detector was used. The unit-cell parameters were refined on the basis of the 28 Bragg reflections, confirming a triclinic lattice, as previously observed by X-ray data (Table 3). A second set of intensity data were collected at higher $\sin(\theta)/\lambda$ up to 0.89 Å⁻¹ with a wavelength of the incident beam of 0.7925(5) Å, focusing on those reflections for which significant intensity could be expected from preliminary results of the first data set. A total of 6525 reflections were collected up to $d_{\min} = 0.56$ Å (with $-12 \le h \le +12$, $-16 \le k \le +16$ and $-16 \le l \le +16$, Table 3), using pure ω -scan, ω - θ scan and ω - 2θ scan strategy (Table 3), out of which 5440 with $F_{\rm O} > 4\sigma(F_{\rm O})$

(Table 3). Integrated intensities were then corrected for the Lorentz effect; absorption correction was found to be negligible. Further details pertaining to the data collection are reported in Table 3.

Mössbauer spectroscopy

Transmission ⁵⁷Fe Mössbauer spectra were collected at room temperature using a Mössbauer apparatus (HALDER electronics, Germany) in horizontal arrangement (⁵⁷Co/Rh single line thin source, constant acceleration mode, symmetric triangular velocity shape, multichannel analyser **Table 4** Fractional atomiccoordinates and displacementparameters ($Å^2$) of ferroaxinitebased on the neutron structurerefinement

Site	s.o.f.	x/a	y/b	z/c	$U_{ m eq}$
T1	Si 1	0.21088(10)	0.45011(8)	0.23474(9)	0.0051(1)
T2	Si 1	0.21938(10)	0.27470(8)	0.52352(8)	0.0045(1)
Т3	Si 1	0.69910(10)	0.25627(8)	0.01159(8)	0.0051(1)
T4	Si 1	0.64123(10)	0.01939(8)	0.23025(8)	0.0046(1)
T5	B 1	0.46163(8)	0.63455(6)	0.28663(6)	0.0041(1)
X1	Ca 1	0.74635(10)	0.34819(8)	0.39532(8)	0.0083(2)
X2	Ca 0.990(2), Mn 0.010(2)	0.18262(10)	0.10048(8)	0.08406(8)	0.0088(2)
Y	Mn 0.241(1), Fe 0.759(1)	0.76734(7)	0.59191(6)	0.11238(6)	0.0090(1)
Z1	Al 0.947(3), Fe 0.053(3)	0.05229(11)	0.80096(9)	0.25382(9)	0.0047(2)
Z2	Al 1	0.35219(12)	0.93630(10)	0.42114(10)	0.0044(1)
01	01	0.05439(8)	0.60314(6)	0.18978(6)	0.0073(1)
02	01	0.23147(8)	0.33836(7)	0.09633(7)	0.0098(1)
03	01	0.41890(7)	0.48708(6)	0.31239(7)	0.0074(1)
04	01	0.13602(8)	0.37268(7)	0.37032(7)	0.0107(1)
05	01	0.02202(7)	0.24227(6)	0.56404(6)	0.0072(1)
06	01	0.32690(7)	0.37996(6)	0.64496(6)	0.0068(1)
07	01	0.38066(7)	0.12749(6)	0.49608(6)	0.0060(1)
08	01	0.53638(8)	0.34365(6)	0.87694(6)	0.0074(1)
09	01	0.87613(7)	0.15473(6)	0.93409(6)	0.0067(1)
010	01	0.76849(8)	0.36671(7)	0.13903(7)	0.0093(1)
011	01	0.60328(8)	0.13526(7)	0.08758(7)	0.0099(1)
012	01	0.43639(7)	0.98157(6)	0.24383(6)	0.0066(1)
013	01	0.72103(8)	0.09950(6)	0.38433(6)	0.0067(1)
014	01	0.79380(7)	0.87452(6)	0.17720(6)	0.0077(1)
015	01	0.32548(7)	0.74608(6)	0.35443(6)	0.0055(1)
016	01	0.09649(7)	0.99606(6)	0.32272(6)	0.0064(1)
Н	H 1	-0.00645(16)	1.04412(14)	0.37875(14)	0.0216(2)

 U_{eq} is defined as one-third of the trace of the orthogonalized U_{ii} tensor

with 1024 channels, velocity scale calibrated to α -iron). For Mössbauer absorber preparation, 68 mg of the sample were first ground in ethanol, then filled into Cu rings (inner diameter 10 mm and covered with a high-purity Al foil on one side) and mixed with epoxy resin to fix the sample. The folded spectra were analysed using classical full static Hamiltonian site analysis (using Lorentzian-shaped doublets) with the program RECOIL (Rancourt and Ping 1991).

Results: neutron structure refinement of axinite

The neutron intensity data collected at 293 K were first processed in order to calculate the normalized structure factors (*E*'s) and their statistical distributions (using the program E-Statistics, implemented in the WinGX package; Farrugia 1999). The structure was found to be centrosymmetric (with 89 % likelihood; $|E^2 - 1| = 0.913$). The anisotropic

structure refinement was then performed in the space group *P*-1 using the SHELX-97 software (Sheldrick 1997, 2008), starting from the (H-free) structure model of Andreozzi et al. (2004). The neutron scattering lengths of Ca, Mn, Fe, Al, B, Si, O and H were used according to Sears (1986). The effects of secondary isotropic extinction were corrected following the Larson's formalism (Larson 1967), as implemented in SHELX-97. In order to have a clear picture of the cations distribution among the octahedral and tetrahedral sites, the first cycles of the refinement were conducted with the following protocol:

- The *T1*, *T2*, *T3* and *T4* sites were first modelled with a mixed neutron scattering length of Si and Al and then with Si and B. However, the potential fraction of Al or B was found being not significant (Table 4).
- 2. The *T5* site was modelled with a mixed (Si + B) scattering length. However, the fraction of Si was found being not significant (Table 4).



Fig. 2 Difference-Fourier maps of the nuclear density $(fm/Å^3)$ of ferroaxinite structure at $z \sim 0.38$, phased without the H site [section parallel to (001)]

- 3. The X1 and X2 sites were first modelled with a mixed (Ca + Mn) scattering length. Only at the X2 sites the fraction of Mn was found being significant (Table 4).
- 4. The Y site was modelled with a mixed (Mn + Fe) scattering length, and the Mn/Fe fraction was refined (Table 4).
- 5. The Z1 and Z2 sites were modelled with a mixed (Al + Fe) scattering length. Only the Z1 site showed a significant fraction of Fe (Table 4).
- 6. All the oxygen sites (i.e. *O1–O16*, Table 4) were modelled with scattering length of O, showing full site occupancies.

When convergence was achieved, one negative residual peaks was found in the final difference-Fourier map of the nuclear density at $x \sim -0.006$, $y \sim 1.04$, $z \sim 0.38$ (Fig. 2). The final cycles of anisotropic refinement were then conducted assigning H to this residual peak (as hydrogen has a negative neutron scattering length), which was modelled with an anisotropic displacement regime, and its site occupancy factor was refined, giving a final full occupancy within 1σ . Convergence was rapidly achieved, with no significant correlation among the refined parameters in the variance–covariance matrix. No peaks larger than -0.86/+0.87 fm/Å³ were found in the final difference-Fourier map of the nuclear density at the end of the refinement (Table 3). The final agreement index $R_1(F)$ was 0.032 for 251 refined parameters and 5440 unique reflections with $Fo > 4\sigma(Fo)$ (Table 3). Atomic coordinates and anisotropic displacement parameters are listed in Tables 4 and 5; relevant bond lengths and angles are listed in Table 6.

Results: Mössbauer spectrum of axinite

The ⁵⁷Fe Mössbauer spectrum of axinite is shown in Fig. 3. It is dominated by a doublet with hyperfine parameters typical for Fe^{2+} in octahedral coordination. There is no evidence of a second Fe²⁺ doublet as found, e.g., by Andreozzi et al. (2004) in some rare cases. A distinct shoulder at ~0.4 mm/s is indicative for the presence of Fe^{3+} ; the obtained hyperfine parameters (Table 7) are typical for Fe^{3+} in octahedral coordination, and the relative area fraction of this component is 10.3(9) %. The width of the shoulder for Fe^{3+} is small, and, as for Fe^{2+} , no evidence is seen for a second Fe³⁺ component, which would indicate that Fe³⁺ is distributed over different sites. The obtained spectrum for axinite of this study and the derived ⁵⁷Fe Mössbauer parameters agree very well with data previously reported in the literature (e.g. Fuchs et al. 1997; Andreozzi et al. 2004; Zabinski et al. 2002; Filip et al. 2006).

Discussion

The chemical composition of the axinite from Colebrook Hill, Rosebery district, Tasmania, Australia, based on EPMA-WDS and ICP-AES analyses, suggests that the sample is a ferroaxinite (following Andreozzi et al. 2004). EPMA-WDS and ICP-AES data are mutually consistent. The ICP-AES data are considered as the reference ones, and the re-calculation of the crystal-chemical formula in atoms per formula unit (a.p.f.u.) in Table 2 is based on the assumption that $Fe^{2+}/Fe^{3+} = 9:1$, as deduced by the Mössbauer data (Table 7). In addition, the ⁵⁷Fe Mössbauer spectrum of our ferroaxinite shows unambiguously the occurrence of Fe^{2+} and Fe^{3+} in octahedral coordination only. The narrow width of the peaks suggests ruling out a potential multisites distribution of Fe^{3+} . In this light, it is highly likely that Fe^{3+} populates the octahedral Z1 site only, along with Al, according to the previous experimental findings (e.g. Andreozzi et al. 2000a, b, 2004; Zabinski et al. 2002; Filip et al. 2006). We tend to consider unlikely the assumption that Fe³⁺ populates the tetrahedral sites, as previously reported by Pieczka and Kraczka (1994) and Fuchs et al. (1997) for samples of Polish and Norwegian axinites, respectively.

The neutron structure refinement of ferroaxinite of this study provides a structure model in general agreement with the experimental findings so far reported:

Table 5Refineddisplacement parameters		U_{11}	U ₂₂	U ₃₃	U ₂₃	<i>U</i> ₁₃	<i>U</i> ₁₂
$(Å^2)$ in the expression:	Tl	0.0055(3)	0.0053(3)	0.0046(3)	0.0011(2)	0.0005(2)	-0.0013(2)
$-2\pi^{2}[(ha^{*})^{2}U_{11} + \cdots + 2hka^{*}b^{*}U]$	<i>T</i> 2	0.0040(2)	0.0049(3)	0.0047(3)	0.0010(2)	0.0006(2)	-0.0012(2)
$+ 2klb^*c^*U_{23}$	<i>T3</i>	0.0058(3)	0.0051(3)	0.0042(3)	0.0009(2)	0.0013(2)	-0.0003(2)
2.0-	<i>T4</i>	0.0050(3)	0.0062(3)	0.0029(3)	0.0007(2)	0.0009(2)	-0.0017(2)
	<i>T4</i>	0.0041(2)	0.0045(2)	0.0039(2)	0.0002(2)	0.0012(2)	-0.0006(2)
	X1	0.0097(3)	0.0062(3)	0.0081(3)	0.0018(2)	-0.0005(2)	-0.0007(2)
	X2	0.0101(3)	0.0090(3)	0.0070(3)	0.0021(2)	-0.0023(2)	-0.0037(2)
	Y	0.0062(2)	0.0130(2)	0.0081(3)	0.0036(2)	0.0008(2)	-0.0022(2)
	Z1	0.0048(3)	0.0050(3)	0.0045(4)	0.0004(3)	0.0008(2)	-0.0012(2)
	Z2	0.0043(3)	0.0049(3)	0.0040(4)	-0.0005(3)	0.0007(2)	-0.0009(2)
	01	0.0072(2)	0.0057(2)	0.0084(2)	0.0012(2)	0.0001(2)	-0.0005(2)
	02	0.0127(2)	0.0082(2)	0.0082(2)	-0.0019(2)	0.0019(2)	-0.0020(2)
	03	0.0060(2)	0.0064(2)	0.0098(2)	0.0017(2)	-0.0002(2)	-0.0020(2)
	04	0.0096(2)	0.0146(2)	0.0085(2)	0.0072(2)	-0.0002(2)	-0.0040(2)
	05	0.0055(2)	0.0113(2)	0.0054(2)	0.0017(2)	0.0012(2)	-0.0026(2)
	06	0.0055(2)	0.0071(2)	0.0076(2)	-0.0018(2)	-0.0001(2)	-0.0018(2)
	07	0.0049(2)	0.0055(2)	0.0074(2)	-0.0013(2)	0.0004(2)	-0.0010(2)
	08	0.0076(2)	0.0091(2)	0.0046(2)	0.0020(2)	0.0013(2)	0.0007(2)
	09	0.0062(2)	0.0072(2)	0.0066(2)	0.0004(2)	0.0020(2)	-0.0004(2)
	010	0.0117(2)	0.0100(2)	0.0064(2)	-0.0014(2)	0.0007(2)	-0.0031(2)
	011	0.0113(2)	0.0117(2)	0.0079(2)	0.0050(2)	0.0011(2)	-0.0039(2)
	012	0.0060(2)	0.0097(2)	0.0051(2)	0.0006(2)	0.0014(2)	-0.0033(2)
	013	0.0078(2)	0.0085(2)	0.0049(2)	-0.0011(2)	0.0018(2)	-0.0036(2)
	014	0.0055(2)	0.0093(2)	0.0073(2)	-0.0010(2)	0.0004(2)	0.0003(2)
	015	0.0055(2)	0.0056(2)	0.0053(2)	-0.0007(2)	0.0015(2)	-0.0011(2)
	016	0.0054(2)	0.0072(2)	0.0066(2)	-0.0004(2)	0.0009(2)	-0.0012(2)
	<i>H</i>	0.0186(5)	0.0223(5)	0.0239(6)	-0.0038(4)	0.0101(4)	-0.0004(4)

site positions, bond distances and angles are comparable with those previously published (e.g. Takéuchi et al. 1974; Swinnea et al. 1981; Belokoneva et al. 1997, 2001; Andreozzi et al. 2000a, b, 2004; Filip et al. 2006). In our structure refinement, the tetrahedral T1, T2, T3 and T4 sites are fully occupied by Si, whereas the T5 site is fully occupied by B. There is no evidence of Si at the T5, or Al or Fe^{3+} at the *T1–T5* sites. The structural and chemical data of this study suggest that the amount of B in ferroaxinite is that expected from the ideal stoichiometry: 2 a.p.f.u. (for 32 O). The T5 tetrahedron shows an average intra-tetrahedral bond distance of ~1.485 Å, with a difference between the longest and the shortest bond distance of about 0.1 Å (Table 6). In our refinement, the X1 site is found to be fully populated and the X2 mainly populated by Ca, along with a low fraction of Mn (Table 4). The octahedral Y site is mainly populated by Fe (i.e. we expect Fe²⁺, ~76 %) and in minor fraction by Mn (~24 %). The Z1 site is almost fully occupied by Al (i.e. ~95 %), and only a minor fraction of Fe (we expect Fe^{3+} , ~5 %) is present. The Z2 site contains Al only.

The chemical formula deduced on the basis of the structure refinement is the following:

$${}^{X1,X2}(\text{Ca}_{3.98}\text{Mn}_{0.02})_{\varSigma 4}{}^{Y}(\text{Mn}_{0.48}\text{Fe}_{1.52})_{\varSigma 2} \\ {}^{Z1,Z2}(\text{Fe}_{0.10}\text{Al}_{3.90})_{\varSigma 4}{}^{T1,T2,T3,T4}\text{Si}_{8}{}^{T5}\text{B}_{2}\text{O}_{30}(\text{OH})_{2},$$

whereas that obtained by ICP-AES is:

$$\begin{split} & {}^{X1,X2}\text{Ca}_{4.03}^{Y}\left(\text{Mn}_{0.42}\text{Mg}_{0.23}\text{Fe}_{1.39}^{2+}\right)_{\varSigma 2.04} \\ & {}^{Z1,Z2}\left(\text{Fe}_{0.15}^{3+}\text{Al}_{3.55}\text{Ti}_{0.12}\right)_{\varSigma 3.82} {}^{T1,T2,T3,T4}(\text{Ti}_{0.03}\text{Si}_{7.97})_{\varSigma 8} \\ & \text{B}_{1.96}\text{O}_{30}(\text{OH})_{2.18}. \end{split}$$

As Mg and Ti were not considered in the neutron structure refinement, a comparison between the results of the refinement and the chemical analysis should be done on the basis of the number of electrons per site:

Table 6 Relevant bond distances (Å) and angles (°) based on the neutron structure refinement of ferroaxinite

<i>T1-</i>	02	1.5885(10)	X2-	012	2.2428(9)
	01	1.6169(9)		O2	2.2883(10)
	O4	1.6369(9)		O9	2.3645(9)
	03	1.6527(9)		O14	2.3929(9)
	Av.	1.6238		O9	2.4718(9)
				O16	2.5776(9)
T2-	05	1.5995(9)		[011	2.9030(10)]
	07	1.6133(9)		[011	3.0949(9)]
	O4	1.6311(9)			
	O6	1.6546(10)			
	Av.	1.6246	<i>Y</i> -	O2	1.9979(8)
				O10	2.0915(8)
ТЗ-	O10	1.6034(10)		01	2.0976(8)
	09	1.6291(9)		08	2.1407(7)
	011	1.6406(9)		O6	2.3577(8)
	08	1.6453(9)		O14	2.6880(8)
	Av.	1.6296			
T4-	O12	1.6021(9)	Z1-	O14	1.8634(9)
	O14	1.6328(9)		05	1.8675(10)
	013	1.6347(9)		01	1.8890(10)
	011	1.6443(9)		O9	1.9096(10)
	Av.	1.6285		O16	1.9568(10)
T5-	015	1.4367(8)		015	1.9965(9)
	03	1.4868(8)			
	08	1.4882(8)	Z2-	012	1.8675(10)
	O6	1.5292(8)		O15	1.8727(11)
	Av.	1.4852		O16	1.8868(10)
				07	1.9058(11)
X1-	O13	2.3330(9)		07	1.9194(10)
	O10	2.3386(10)		O13	1.9482(10)
	05	2.3509(9)			
	O3	2.4352(9)			
	O6	2.4675(9)			
	015	2.5872(9)			
	O4	2.8897(9)			
016–H	H 0.970(1)				
016–H	a 0.991(1))			
016	05 3.096(1)			
<i>H</i> … <i>O</i> 5	2.450(1)				
016–H	<i>I</i> 05 123	.9(1)			
016	013 2.777	(1)			
H···Ol	3 1.914(1))			
016–H	<i>IO13</i> 14	6.9(1)			

Phys Chem Minerals (2016) 43:341-352



Fig. 3 ⁵⁷Fe Mössbauer spectrum of ferroaxinite at room temperature

showing a very good agreement.

For the first time, the neutron diffraction data allowed an unambiguous localization of the H site, which forms a hydroxyl group with the oxygen atom at the *O16* site as *donor* (Tables 4 and 6). The H-bonding scheme in the axinite structure is fully shown by Fig. 4: the *O16–H* distance (corrected for riding motion effect, Busing and Levy 1964) is 0.991(1) Å and an asymmetric bifurcated bonding configuration occurs, with *O5* and *O13* as acceptors (Table 6):

- O16…O5 = 3.096(1) Å, H…O5 = 2.450(1) Å and O16− H…O5 = 123.9(1)°;
- $O16\cdots O13 = 2.777(1)$ Å, $H\cdots O13 = 1.914(1)$ Å and $O16-H\cdots O13 = 146.9(1)^{\circ}$.

The *O16–H···O5* bond is weaker than the *O16–H···O13* one, as suggested by the refined distances and angles. Other potential H-bonds, with *O5'* [i.e. *O16···O5* = 2.786(1) Å and *O16–H···O5* = 83(1)°], *O14* [i.e. *O16···O14* = 2.796(1) Å and *O16–H···O14* = 80.1(1)°] and *O16* [i.e. *O16···O16* = 3.634(1) Å and *O16-H···O16* = 134.0(1)°] as acceptors, can be ruled out as energetically unfavourable.

A clue of the H-bonding configuration in axinite structure was first reported by Takéuchi (1975), on the basis of the X-ray and IR data available at that time (Beran 1971; Takéuchi et al. 1974). A more recent study by Filip et al. (2006) based on FTIR and X-ray data, in which the H-bonding scheme in axinite is described, reports: (1) an

 $^{\rm a}$ Corrected for riding motion effect following Busing and Levy (1964)

(SCND)
$${}^{X1,X2}[80.1e-]^{Y,Z1,Z2}[104.8e-]^{T1,T2,T3,T4}[112.0e-]^{T5}[10.0e-]O_{30}(OH)_2,$$

(ICP - AES) ${}^{X1,X2}[80.6e-]^{Y,Z1,Z2}[102.2e-]^{T1,T2,T3,T4}[112.2e-]^{T5}[9.8e-]O_{30}(OH)_{2.18},$

 Table 7
 ⁵⁷Fe Mössbauer parameters of ferroaxinite

IS (mm/s)	QS (mm/s)	FWHM (mm/s)	Area (%)	Site assignment
1.140(5)	2.089(5)	0.302(6)	89.7(8)	Fe ²⁺
0.378(6)	0.385(8)	0.286(5)	10.3(9)	Fe ³⁺

IS isomer shift, QS quadrupole splitting with respect to α -iron at room temperature, FWHM full width at half maximum



Fig. 4 H-bonding scheme (with a bifurcated configuration) in axinite structure based on the neutron structure refinement of this study. Atomic displacement probability factor: 50 %

intense IR-active band (broad and asymmetric) with the maximum at ~ 3382 cm⁻¹, along with a very low-intensity band at ~3485 cm⁻¹, and (2) a hydrogen bonding scheme based on $O16-H \sim 0.83$ Å, $O16...O13 \sim 2.78$ Å and $O16-H...O13 \sim 151^{\circ}$. The authors considered a H-bond configuration based on one acceptor only (i.e. 013) and deduced from the distance-frequency correlation of Libowitzky (1999) the expected $O_{\text{donor}} \cdots O_{\text{acceptor}}$ distance of 2.8 Å, in good agreement with the refined $O_{\text{donor}} \cdots O_{\text{acceptor}}$ distance (Filip et al. 2006). The low-intensity band at \sim 3485 cm⁻¹ was ascribed to *O16* as potential acceptor. with $O16...O16 \sim 3.11$ Å and $O16-H...O16 \sim 126^{\circ}$. The neutron structure refinement of this study shows that the H-bonding scheme in axinite structure is actually more complex: the broad and asymmetric IR band with the maximus at $\sim 3382 \text{ cm}^{-1}$ might represent the convolution of at least two different contributes, according to the bifurcated H-bonding configuration here found. The low-intensity IR band at $\sim 3485 \text{ cm}^{-1}$ can accordingly be due to energetically weaker H-bonds, with 05', 014 and 016 as acceptors, in agreement with the experimental findings of this study.

A careful inspection of the displacement parameters reported in Table 5 shows a pronounced magnitude of the displacement ellipsoid of H, though with a not drastic anisotropy. However, the difference-Fourier map of the nuclear density, phased without the H site, shows no evidence of site split (Fig. 2). Even the O4 and O11 sites are described with a slightly pronounced anisotropic displacement ellipsoids, which likely reflect their bonding configurations and the multielement population of the octahedral sites, with a resulting local difference in the cation-O bonding distances.

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