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1 **Acid mine drainage sources and hydrogeochemistry at the Yatani mine, Yamagata,**
2 **Japan: A geochemical and isotopic study**

3
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15
16 **1. Introduction**

17 There are more than 5,500 closed or abandoned non-ferrous metal mines in Japan,
18 many of which produce acid mine drainage (AMD; Ueda and Masuda, 2005; Herrera et
19 al., 2007a, b). AMD is not only acidic but may also contain toxic contaminants such as
20 Cu, Pb, Zn, and As, and is a serious environmental problem in many active, closed, and
21 abandoned mines (Younger, 2001; Gault et al., 2005; Molson et al., 2005; Boulabah et
22 al., 2006; Lim et al., 2008; Hien et al., 2012; Hierro et al., 2014; Skierszkan et al., 2016;
23 Zhao et al., 2017).

24 AMD typically forms in old mine workings (mining levels, drifts and shafts), waste-
25 rock dumps, and tailings storage facilities where sulfide minerals such as pyrite are
26 exposed to oxidizing conditions (Singer and Stumm, 1970; Lawson, 1982; Tabelin et
27 al., 2017a, b). The problem is commonly managed by neutralization (Kalin et al., 2006),
28 precipitating heavy metals with basic materials such as limestone, lime, or caustic soda.

29 This method has been effectively used in many closed or abandoned mine sites over the
30 last 40 years, at a total subsidiary aid cost of ~70 billion JPY (~770 million USD)
31 (Kawabe, 2017). Water treatment is not a sustainable solution owing to its continuing
32 high cost. Land surface remediation and wetland treatment have been implemented in
33 Japan (Tomiyaama et al., 2016a; Ogino, 2017; Okimoto et al., 2017), but again the cost
34 is unsustainable. Now, a “green remediation of abandoned mines” program is to be
35 implemented with the goal of returning mine landscapes to their pre-mining state, while
36 also reducing treatment costs with a variety of techniques including the sealing of
37 mining levels and wetland treatment (Nagai, 2018). An understanding of the origin and
38 flow of groundwater in abandoned mines is essential for the development of
39 countermeasures such as back-filling and sealing of surfaces in recharge areas.

40 Groundwater flow modeling, based on the monitoring of changes in geochemistry of
41 water percolating into mines, has provided useful information in the development of an
42 AMD mitigation strategy (Tomiyaama et al., 2010a). Previous studies have characterized
43 geochemical and isotopic compositions of AMD in Japan (Iwatsuki and Yoshida, 1999;
44 Okumura, 2003; Mahara et al., 2006; Tomiyaama et al., 2010b; Yamaguchi et al., 2015;
45 Tomiyaama et al., 2016b) and other countries (Wunsch et al., 1999; Hazen et al., 2002;
46 Sracek et al., 2004; Casiot et al., 2005; Lim et al., 2008; Gammons et al., 2010; Pabst et
47 al., 2018; Ramasamy et al., 2018). Further information could be provided by studies of
48 the origin and hydrogeochemistry of AMD in mining areas where water samples can be
49 collected from inside mining levels and surface sites.

50 The Yatani mine area in central Honshu Island, Japan, was selected for this study.
51 The area includes a closed non-ferrous sulfide mine that had been producing galena,
52 sphalerite, native Au, and argentite. The objectives of the study were to evaluate the

53 geological and hydrological characteristics of the mine, to identify important minerals,
54 and to determine their roles in the formation of mine drainage (AMD and neutral mine
55 drainage). Groundwater flow systems in the mine area were evaluated through water
56 quality and isotopic studies interpreted by principal component analysis (PCA).

57

58 **2. Materials and methods**

59 *2.1 Geology and history of the Yatani mine*

60 The Yatani mine area is located 28 km southwest of Yonezawa City, within the
61 Bandai–Asahi National Park (Fig. 1). It is situated in the green tuff area of northeast
62 Japan, and includes a granitic basement overlain by Tertiary–Miocene volcanic and
63 sedimentary rocks. Surface mining began in 1870 after the discovery of Au ore
64 outcrops. When these were exhausted, mining continued underground for several
65 decades. Taihei Mining Co. Ltd acquired mining rights to the area in 1952 and began
66 mining Pb and Zn ores. By 1978 the monthly crude ore output was >10,000 tons with
67 average Pb and Zn grades of 2% and 4.4%, respectively. Decreasing metal prices and
68 the appreciating Japanese yen eventually made the operation unprofitable, and it closed
69 in 1988 (Sato et al., 1978).

70 The Yatani mineral deposit is of hydrothermal origin, formed within green tuff host
71 rock. It includes two types of veins: one rich in Pb and Zn, and the other containing Au
72 and Ag. The Pb–Zn ore deposits constitute the main veins of the mine, reaching a
73 maximum width of ~10 m and closely following NE–SW- and E–W-trending faults in
74 the area. The Pb–Zn veins comprise mainly quartz, galena, sphalerite, pyrite, calcite,
75 and chlorite and sometimes include native Au and argentite. The Au–Ag ore veins
76 contain quartz, native Au, argentite, pyrite, and rhodochrosite. In both types of veins,

77 ore minerals exhibit striped textures, indicating that quartz crystallized first, followed by
78 precipitation of galena and zinblende, and finally by deposition of native Au and
79 argentite (Taniguchi, 1969).

80 Two mining methods were employed at the mine, involving the cut-and-fill method
81 in deeper parts using sand, slime, and tailings as filling materials that did not
82 consolidate with cement; and the shrinkage-stoping method in shallower parts. A
83 mineral concentration system with its own power substation and repair facilities was
84 constructed underground. The adit mouth of the drainage level (level -2L) is located at
85 elevations of 626 and 2,280 m from the shaft. The tunnel extends from the bottom level
86 (-10 L) to the top (+2 L) level with a height difference of 600 m and runs E-W for ~2.5
87 km. Old mine workings deeper than the drainage level (level -2L) are kept submerged,
88 and AMD from the drainage level is presently treated by neutralization.

89

90 *2.2 Water sample collection and analysis*

91 Water samples were collected on 9 and 10 November 2008, from locations shown in
92 Figs 1 and 2. In the old mine workings (mining levels, drifts, and a shaft), samples were
93 collected from the shaft and other prominent parts of mine workings where significant
94 discharge of groundwater was observed, for example, from cracks in drainage level
95 walls, and fallen water from the back of the drift and the top of the shaft. Samples were
96 also collected from several points in major rivers and streams traversing the mining
97 area. All samples were stored in 1 L polypropylene bottles before filtering through 0.22
98 µm membrane filters. Temperature, pH, electrical conductivity (EC), and oxidation-
99 reduction potential (ORP) were measured on-site using a thermo-pH meter (KS-701,
100 Shindengen Electric Manufacturing Co. Ltd., Japan), an EC meter (B-173, Horiba Ltd.,

101 Japan), and an ORP meter (RM-12P, DKK-TOA Co., Japan), respectively. The ORP
102 meter was equipped with an Ag–AgCl electrode, and measured values were converted
103 to standard redox potentials (Eh) by adding a temperature-dependent constant for the
104 electrode solution. The AMD flux on 9 November 2008, when mine drainage samples
105 were taken, was $2.58 \text{ m}^3 \text{ min}^{-1}$. There was no rainfall on 10 November 2008 when river
106 water was sampled; precipitation during the 30 days before sampling total 110 mm (Fig.
107 3).

108 All sample analyses were performed at the Analytical Center of Mitsubishi
109 Materials Techno Corporation. Major anions (Cl^- and SO_4^{2-}) were analyzed by ion
110 chromatography using Dionex DX-120 and IonPac As14 resin columns (Nippon Dionex
111 K.K., Japan). Major cations (Na^+ , K^+ , Ca^{2+} , and Mg^{2+}), total Fe, and total Mn were
112 analyzed by either inductively coupled plasma mass spectrometry (ICP–MS; Agilent
113 7700X, Agilent Technologies Japan Ltd., Japan) or inductively coupled plasma atomic
114 emission spectroscopy (ICP–AES; ICAP-575, Thermo Fisher Scientific K.K., Japan).
115 Alkalinity was determined by titration with 0.01 M H_2SO_4 to pH 4.8. Charge balances
116 were within $\pm 5\%$, indicating that all major cations and anions were adequately
117 quantified. Analytical uncertainties of ICP–MS, ICP–AES, and ion chromatography
118 results were all 2%–5%.

119 Oxygen isotopic ratios were determined by the CO_2 equilibration method (Epstein
120 and Mayeda, 1953) and hydrogen isotopic ratios by the method of Coleman et al.
121 (1982). $\delta^{18}\text{O}$ ($1000((^{18}\text{O}/^{16}\text{O})_{\text{sample}}/(^{18}\text{O}/^{16}\text{O})_{\text{VSMOW}}) - 1$), ‰) and δD
122 ($1000((^2\text{H}/^1\text{H})_{\text{sample}}/(^2\text{H}/^1\text{H})_{\text{VSMOW}}) - 1$), ‰) values were calculated relative to Vienna
123 Standard Mean Ocean Water (VSMOW) with analytical precisions of 0.1‰ and 1‰,
124 respectively.

125

126 *2.3 Principal component analysis*

127 Principal component analysis (PCA) is a multivariate statistical technique used to
128 reduce complex high-dimensional data sets by converting observations into sets of
129 values or “principal components” (PC) related to original system components (Abdi and
130 Williams, 2010). The approach is ideal for geochemical surveys because each sample is
131 described by multiple factors such as pH, Eh, EC, and ion concentrations. PCA has been
132 applied in many studies comparing the qualities of groundwater, spring water, and river
133 water, and also in the classification of hot-spring waters (Masumoto and Hibiya, 1996;
134 Takamatsu, 1986; Mahlknecht et al., 2003).

135 Here, PCA was performed using the multivariate analysis software Pirouette LE ver.
136 3.11 (GL Sciences Inc., Japan), and it was applied to four variables measured on site
137 (temperature, pH, EC, and Eh), the concentrations of seven major ions (Na^+ , K^+ , Ca^{2+} ,
138 Mg^{2+} , Cl^- , HCO_3^- , and SO_4^{2-}), and the isotopic ratios δD and $\delta^{18}\text{O}$. For normalization of
139 variables, logarithms were used for ion concentrations, and measured values for the four
140 factors measured on site, and isotopic ratios were the antilogs of the measured values.

141

142 **3. Results and discussion**

143 *3.1 Geochemical properties of water samples*

144 The geochemical and isotopic properties of water samples collected from the Yatani
145 mine area are summarized in Table 1 and their corresponding water chemistry displayed
146 in Stiff (Figs 1 and 2) and Piper diagrams (Fig. 4).

147

148 *3.1.1 River water*

149 The pH, EC, Eh, and temperature of river water samples are in the ranges of 6.0–
150 8.9, 1.8–9.1 mS m⁻¹, +353 to +579 mV, and 4.6°C–7.9°C, respectively. Concentrations
151 of major ions vary with location, with Ca²⁺, Na⁺, and HCO₃⁻ being dominant
152 downstream in the Yatanizawa River and the Tengusawa River (Na⁺–HCO₃⁻ or
153 Ca²⁺–HCO₃⁻ type waters (samples R-1, R-2, R-9, R-10, R-11, R-12, R-13, and R-14),
154 while SO₄²⁻ predominates in samples (Ca²⁺–SO₄²⁻ type water; samples R-3, R-4, R-5,
155 R-6, R-7, and R-8) collected upstream where mineralized zones and veins are located
156 (Sato et al., 1978).

157

158 *3.1.2 Mine drainage*

159 The pH, EC, Eh, and temperature of mine drainage are in the ranges of 2.9–7.8,
160 16.5–381 mS m⁻¹, +304 to +656 mV, and 11.6°C–18.7°C, respectively. Mine drainage
161 samples are Ca²⁺–SO₄²⁻ type waters, except for sample M-1 (from near the adit mouth),
162 which is a Ca²⁺–HCO₃⁻ type water. Mine drainage samples M-1 to M-14 from old mine
163 workings has lower pH and higher EC values than river water, indicating significant
164 impact of AMD on groundwater quality. EC, pH, and major ion concentrations of mine
165 drainage also varies with location, with EC values increasing with decreasing pH from
166 adit mouth to veins and shaft (Fig. 5). Drainage sample M-12, from the drift, has the
167 highest EC (381 mS m⁻¹) and lowest pH (2.9). Sample M-11, from the shaft, has high
168 EC (235 mS m⁻¹) and a pH of 3.1. A similar trend was observed with concentrations of
169 total Fe and Mn in mine drainage samples. Total Fe and Mn concentrations in samples
170 collected near the adit mouth (samples M-1 to M-9) were <2.5 and 4.2 mg L⁻¹,
171 respectively, while those in samples from around veins, drift, and the shaft (samples M-
172 10 to M-14) were ~80 and ~50 times higher, respectively.

173

174 *3.2 Hydrogen and oxygen isotopic ratios*

175 The δD and $\delta^{18}O$ values of water samples are in the ranges of -67.1‰ to -55.6‰
176 and -10.8‰ to -8.7‰ , respectively, with relationships illustrated in Fig. 6. Isotopic
177 data plot close to the meteoric water line (Craig, 1961; Dansgaard, 1964), and are very
178 similar to those of river water and rainwater collected in Yamagata Prefecture (Mizota
179 and Kusakabe, 1994). The Tengusawa River water samples from higher elevations
180 (~ 930 m; samples R-9 and R-10) have lower $\delta^{18}O$ and δD values, suggesting that the
181 relationship between elevation and isotopic ratios is likely due to isotopic fractionation
182 during precipitation (Dansgaard, 1964; Lawrence and White, 1991; Mizota and
183 Kusakabe 1994). The ranges of $\delta^{18}O$ and δD values in mine drainage approximate those
184 of river water, barring sample R-14. Rainwater in mine drainage is temporally
185 decoupled from that in river water, so isotopic ratios in river water may not display
186 significant seasonal variations, with ratios in mine drainage effectively being averaged
187 over influx and underground flow. The $\delta^{18}O$ and δD values of mine drainage also vary
188 with location (Fig. 7). Isotopic ratios of drainage from the drift and shaft (samples M-10
189 to M-4) are lower than those of samples from the drainage level (samples M-1 to M-9).
190 $\delta^{18}O$ and δD values of mine drainage from the drift and shaft are similar to those of
191 river water from higher elevations (samples R-9 and R-10; Fig. 6). Drainage from the
192 mine drainage level may thus originate in the Yatanizawa River and its tributaries,
193 whereas that from the drift and shaft may originate in river water at higher elevations
194 (>900 m).

195

196 *3.3 Principal component analysis*

197 PCA results are shown in Table 2. Three principal components (PC1, PC2, and PC3)
198 were selected, which together account for more than 82% of the total variability in the
199 dataset. This indicates that the water quality is dominated by these three factors. The
200 first principal component (PC1) has high loadings for Ca^{2+} , Mg^{2+} , and SO_4^{2-} with values
201 of >0.35 , suggesting that the increase in ion concentrations in mine drainage may be
202 attributed to water–mineral interactions within old mine workings. The high SO_4^{2-}
203 loading in PC1 could be explained by the oxidation of pyrite (FeS_2), a common sulfide
204 mineral in metal deposits and altered rocks. Pyrite oxidation is facilitated by Fe^{3+}
205 formed through the oxidation of Fe^{2+} by dissolved oxygen (Singer and Stumm, 1970;
206 Moses et al., 1987; Tabelin et al., 2017a, b). The high Ca^{2+} loading relates to reactions
207 of calcite (CaCO_3) associated with Au–Ag veins and Pb–Zn veins (Taniguchi, 1969),
208 and possibly to interactions of AMD with concrete used in the construction of the
209 underground concentrator and its supporting facilities. The significant Mg^{2+} loading
210 relates to reactions of AMD with the mineral chlorite $(\text{Mg, Fe, Al})_6(\text{Al, Si})_4\text{O}_{10}(\text{OH})_8$
211 associated with Pb–Zn veins (Taniguchi, 1969) or with cementitious materials
212 containing significant amounts of Mg^{2+} .

213 For the second principal component (PC2), there are high loadings for pH and
214 HCO_3^- , with values of 0.40 and 0.52, respectively, as expected because of the well-
215 established inverse relationship between pH and HCO_3^- concentration (Eang et al.,
216 2018a, b).

217 The third principal component (PC3) has high loadings for H and O isotopic ratios
218 (>0.60), possibly reflecting the strong correlation between them in water samples and
219 the effect of isotopic fractionation during precipitation (Figs 6 and 7).

220 The sum of the contributions of PC1, PC2, and PC3 is 82% (Table 2), indicating
221 that the geochemistry of surface water and mine drainage are satisfactorily explained by
222 the three PCs.

223 Scatter plots of PC1 vs. PC2 and PC1 vs. PC3 are shown in Fig. 8a and b,
224 respectively. Two different groups of data are clearly observed in both plots, belonging
225 to the water samples collected from the river and old mine workings, respectively. The
226 river water samples are on the negative side of PC1 whereas the mine drainage samples
227 have positive PC1 loading values. This indicates the possibility of interactions between
228 groundwater and sulfide minerals, sand slime, and tailings back-filled into excavated
229 mine areas.

230 The PC2 coefficients of mine drainage samples range from positive to negative
231 values in the following order: M-13 > M-14 > M-12 > M-11. This shows that the source
232 of mine drainage in drifts (M-11 and M-12) did not come mainly from the shaft (M-13
233 and M-14). It is possible that a large portion of the water from M-11 and M-12
234 originated from the deeper reductive groundwater. This is consistent with the observed
235 flow pattern in which M-11 and M-12 represent water flowing from the back of the
236 drifts, M-13 was the water dripping from the top of the shaft, and M-14 was spring
237 water from the shaft.

238 According to the principal component scores of PC1 vs PC3 (Fig. 8b), two groups
239 of samples can be classified. The group of higher PC1 scores belongs to river water
240 from the elevation of <900 m and mine drainage from the drainage level. On the other
241 hand, the other group represents river water from the elevation of >900 m and mine
242 drainage samples from the drifts and the shaft. These results indicate that the main
243 source of mine drainage from the drainage level is from the Yatanizawa River while

244 mine drainage from the drifts and shaft were derived from surface water with higher
245 elevation than that of the Yatanizawa River. In addition, the board distribution of PC1
246 with narrow PC3 scores on shift samples indicates the progress of water–mineral
247 interactions from the same source of water, along the flow from the top to bottom in the
248 shift, as represented by an arrow in Fig. 8b.

249

250 *3.4 Concept of groundwater flow and discharge*

251 A conceptual model of drainage flow was devised on the basis of results described
252 in sections 3.1–3.3, as shown in Fig. 9. Groundwater infiltrating from the Yatanizawa
253 River and its tributaries above the mine drainage level, and from the mountain slope
254 (elevation > 900 m) north of the mining area, reacts with sulfide minerals, sand slime
255 and tailings back-filled into excavated mine areas releasing SO_4^{2-} , Ca^{2+} , and Mg^{2+}
256 before flowing out of the drainage level. In the PCA study, PC1 is related to the degree
257 of water–rock interaction, PC2 to the contribution of groundwater from the deep
258 reductive environment, and PC3 to correlation between H and O isotopic ratios and
259 isotopic fractionation during precipitation. The total AMD flux measured at the
260 neutralization treatment plant increased during the snowmelt season of March to May,
261 with the pH decreasing over that time (Fig. 3). This suggests that the groundwater flux,
262 infiltrating from the mountain slope to the excavated area and reacting with sulfide
263 minerals, increased during the snowmelt season. It follows that a possible method of
264 reducing AMD formation would be to prevent contact between dissolved oxygen and
265 sulfide minerals by changing the drainage level from the current –2 level to the 0 level,
266 or by filling the shallow underground excavated area with cementitious materials.

267

268 **4. Conclusions**

269 The findings of this geological, hydrological, geochemical, and isotopic study of acid
270 mine drainage at the Yatani mine are as follows.

271 (1) AMD is formed through interactions between groundwater and sulfide minerals,
272 sand slime, and tailings back-filled into excavated mine areas.

273 (2) The groundwater recharge area for AMD formation is located on the mountain slope
274 at altitudes of ~900 m.

275 (3) AMD formation in the drifts and shaft is more pronounced than in the mine drainage
276 level.

277 (4) Based on a conceptual model of groundwater flow at the mine, AMD formation
278 could be reduced by preventing contact between dissolved oxygen and sulfide
279 minerals by increase the drainage level, or by filling shallow underground excavated
280 areas with cementitious materials.

281

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287

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Table 1. Geochemical properties of water samples.

Sample type	Sample	pH	EC (mS m ⁻¹)	Eh (mV)	Temp (°C)	Concentration (mg L ⁻¹)								Stable isotope ratios (‰)		
						Na ⁺	K ⁺	Ca ²⁺	Mg ²⁺	Cl ⁻	HCO ₃ ⁻	SO ₄ ²⁻	T-Fe	T-Mn	δD	δ ¹⁸ O
River water	R-1	8.9	2.6	353	7.9	2.2	0.3	2.5	0.5	1.9	7.8	4.3	<0.1	<0.1	-61.4	-10.1
	R-2	8.5	9.1	392	6.2	2.3	0.6	18	1.3	2.1	35	23	<0.1	<0.1	-61.7	-10.1
	R-3	8.1	3.7	412	5.0	1.6	0.7	4.4	0.7	1.8	4.4	11	<0.1	0.2	-64.2	-10.4
	R-4	6.0	4.0	525	7.1	1.8	1.9	3.6	0.7	2.1	0.7	15	<0.1	0.1	-60.9	-10.2
	R-5	6.4	4.9	472	6.5	2.7	1.0	6.8	0.6	1.7	9.2	16	<0.1	0.2	-60.6	-10.3
	R-6	6.9	6.2	414	4.6	3.2	0.6	9.9	0.9	1.9	14	21	0.1	<0.1	-61.8	-9.7
	R-7	6.7	3.5	579	5.6	2.7	0.8	4.4	0.4	1.6	7.4	11	<0.1	<0.1	-64.1	-10.3
	R-8	6.7	4.2	507	7.3	2.0	1.5	4.8	1.0	2.1	6.6	13	<0.1	<0.1	-67.1	-10.5
	R-9	6.7	2.1	498	5.5	2.5	0.5	2.1	0.3	1.6	11	1.7	<0.1	<0.1	-69.4	-10.8
	R-10	6.7	1.8	497	6.4	2.3	1.0	1.0	0.3	1.7	8.9	1.0	<0.1	<0.1	-67.7	-10.7
	R-11	6.7	2.0	514	6.8	2.4	0.5	1.7	0.4	1.9	8.6	1.8	0.5	0.1	-64.1	-10.1
	R-12	6.9	3.9	511	6.1	3.2	0.6	5.4	0.7	1.8	22	4.4	0.2	<0.1	-69.4	-10.6
	R-13	6.9	3.6	515	6.5	2.3	0.7	5.4	1.1	1.8	13	10	<0.1	<0.1	-65.9	-10.4
	R-14	7.1	5.2	517	6.2	2.6	0.8	7.6	1.5	2.1	18	11	<0.1	<0.1	-55.6	-8.7
	R-15	6.9	4.6	512	6.4	3.2	0.7	6.4	0.5	1.7	14	11	<0.1	<0.1	-65.2	-10.5
Mine drainage from drainage level (-2L)	M-1	7.6	16.5	512	11.6	5.5	0.7	29	1.8	2.4	70	32	<0.1	<0.1	-63.8	-10.2
	M-2	7.4	22.6	516	13.5	6.1	1.0	43	2.2	2.3	78	62	0.2	1.9	-63.9	-10.2
	M-3	7.4	29.2	518	12.6	5.5	1.3	59	2.4	2.2	93	88	0.1	0.4	-64.1	-10.4
	M-4	7.4	32.6	480	12.8	6.2	1.1	66	2.0	2.2	54	130	<0.1	0.3	-63.2	-10.0
	M-5	7.4	37.2	478	12.0	3.3	1.3	73	5.6	2.2	57	160	2.5	4.2	-64.6	-10.6
	M-6	7.8	70.7	477	16.6	6.8	1.0	66	1.6	2.1	53	130	<0.1	1.6	-65.0	-10.6
	M-7	7.6	73.6	439	17.4	18	1.3	120	3.6	2.3	110	260	<0.1	0.5	-66.1	-10.6
	M-8	7.1	97.2	304	17.1	17	1.7	210	8.9	2.3	260	380	<0.1	2.1	-63.7	-9.9
	M-9	7.0	37.5	455	17.2	10	2.0	230	8.4	2.1	220	400	<0.1	<0.1	-63.4	-10.4
	M-10	5.4	70.7	459	18.3	8.3	1.8	340	17	2.2	230	770	22	33	-61.1	-10.0
Mine drainage from drifts	M-11	2.9	381	476	17.8	5.4	4.3	290	85	2.1	0	3200	210	210	-67.1	-10.7
	M-12	3.1	235	452	17.7	6.3	3.3	170	45	2.2	0	1500	47	170	-64.9	-10.7
Mine drainage from shaft	M-13	6.7	32.8	432	16.8	15	0.4	39	1.0	2.0	61	80	1.4	0.3	-66.7	-10.8
	M-14	5.6	127	656	18.7	14	2.7	210	19	1.9	34	810	89	62	-66.1	-10.8

Table 2. Factor loadings and contributions of principal components (PCs).

Variables		PC1	PC2	PC3
pH		-0.26	0.40	-0.01
EC		0.33	-0.19	-0.12
Eh		0.12	-0.46	-0.03
Temperature		0.36	0.23	-0.10
Chemical composition	Na ⁺	0.29	0.33	-0.12
	K ⁺	0.31	-0.21	0.12
	Ca ²⁺	0.35	0.26	0.06
	Mg ²⁺	0.39	0.04	0.08
	Cl ⁻	0.24	0.12	0.34
	SO ₄ ²⁻	0.38	0.11	0.12
	HCO ₃ ⁻	-0.12	0.52	-0.02
Stable isotopes	δ D	-0.06	-0.08	0.66
	δ^{18} O	-0.11	0.09	0.60
Rate of contribution (%)		46.3	23.1	13.2

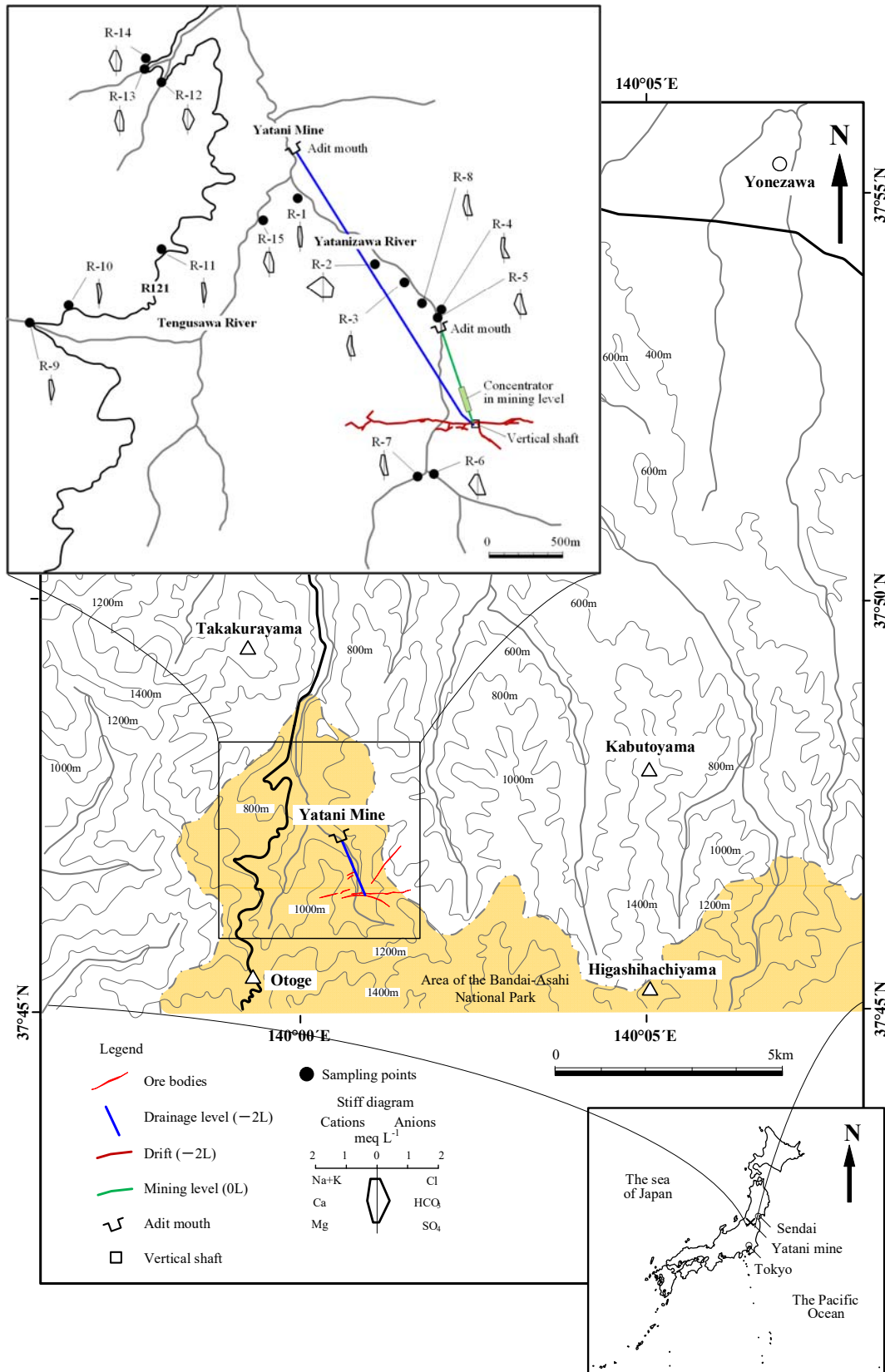


Fig. 1. Location of water sampling points, with corresponding water sample properties indicated by Stiff diagrams.

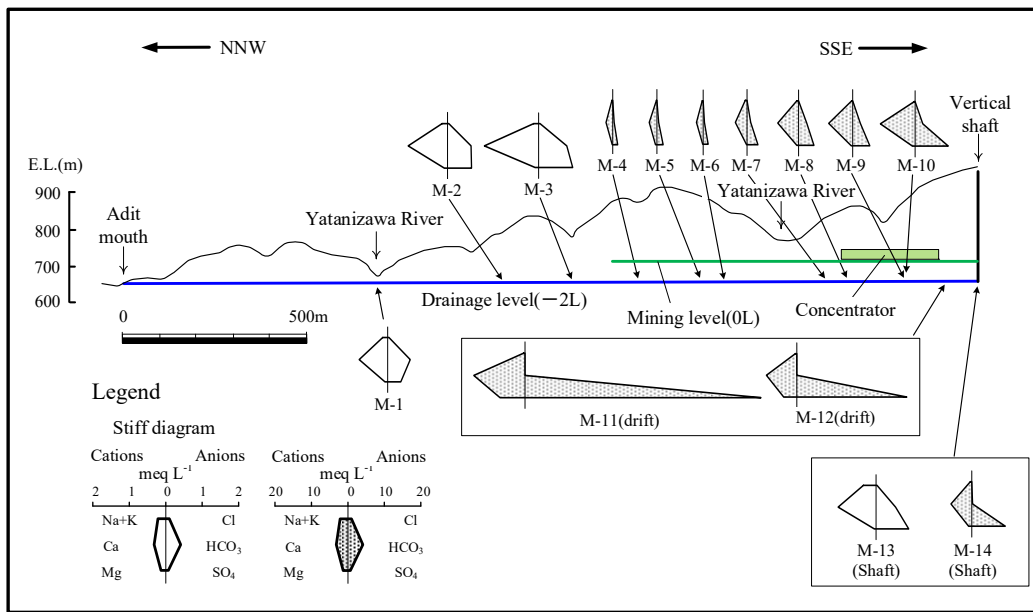


Fig. 2. Stiff diagrams of mine drainage samples from the drainage level (-2L), including samples collected from the drift and shaft (note that the concentrations in shaded Stiff diagrams are 10 times higher than those in unshaded diagrams).

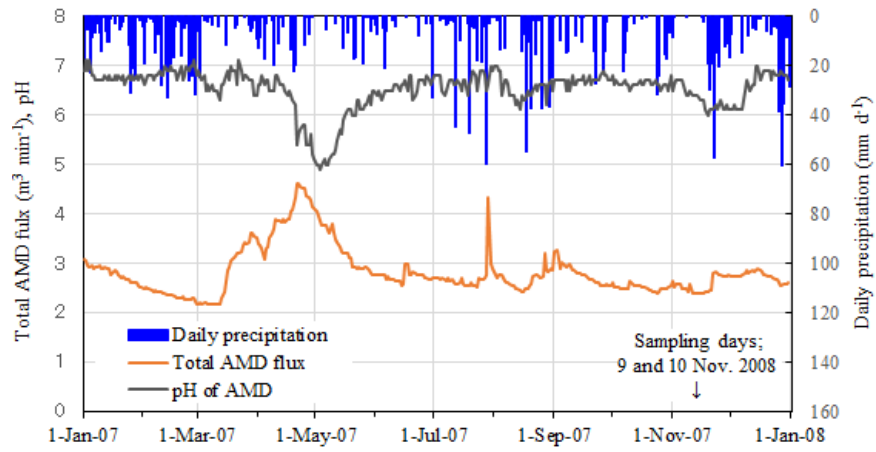


Fig. 3. Relationships between precipitation, total AMD flux, and pH measured at the neutralization treatment plant.

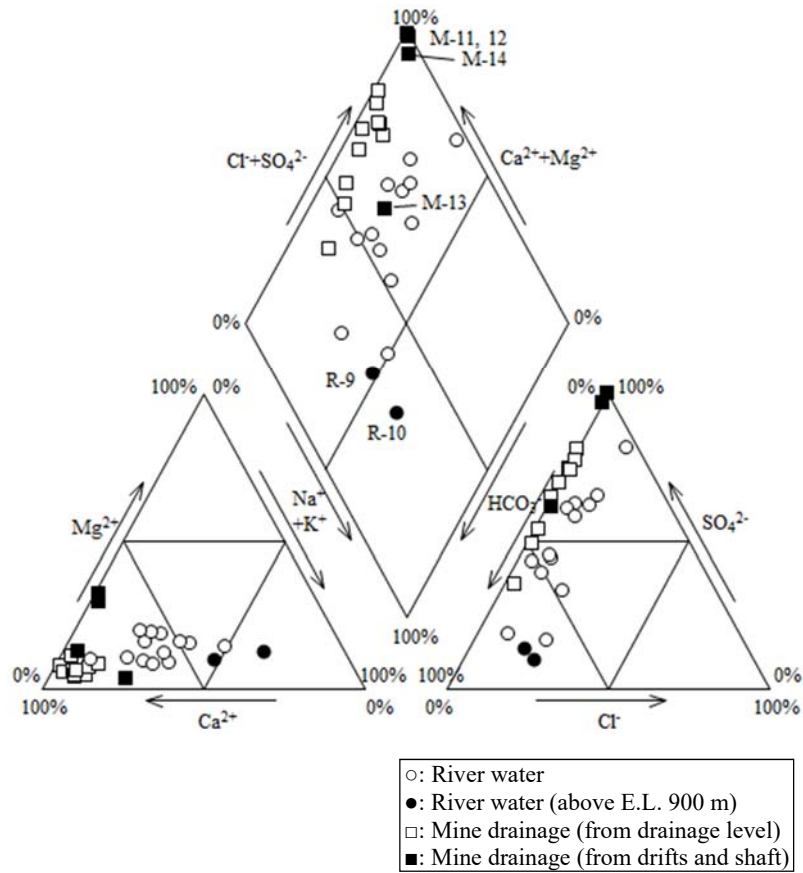


Fig. 4. Piper diagram showing the hydrochemical facies of river water, mine drainage, and AMD samples collected from the Yatani mine area.

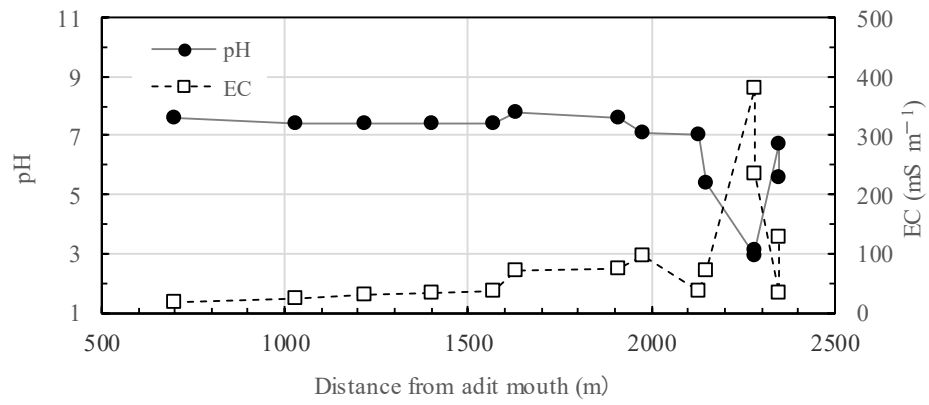


Fig. 5. Spatial variations in pH and EC of mine drainage samples along the mine drainage level.

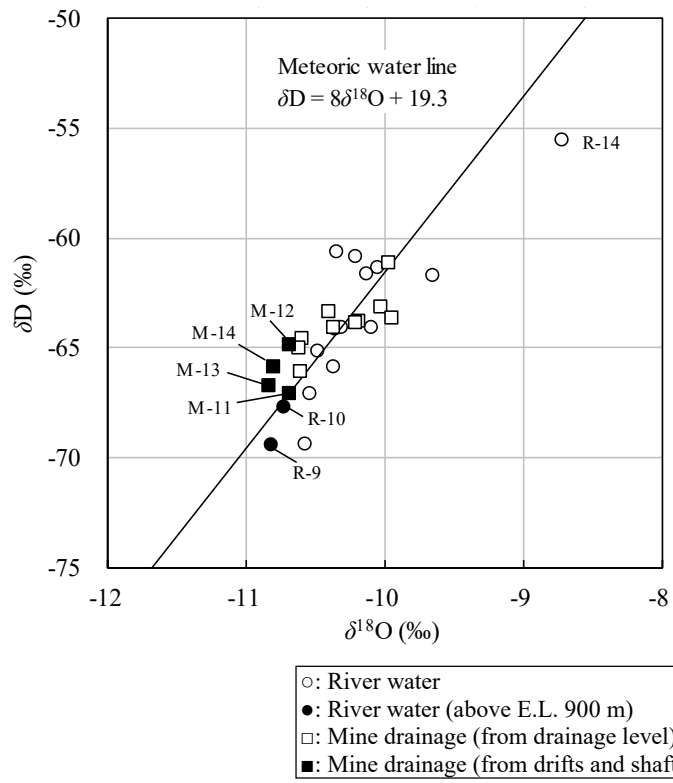


Fig. 6. Relationship between δD and $\delta^{18}O$ values of water samples, relative to the meteoric water line.

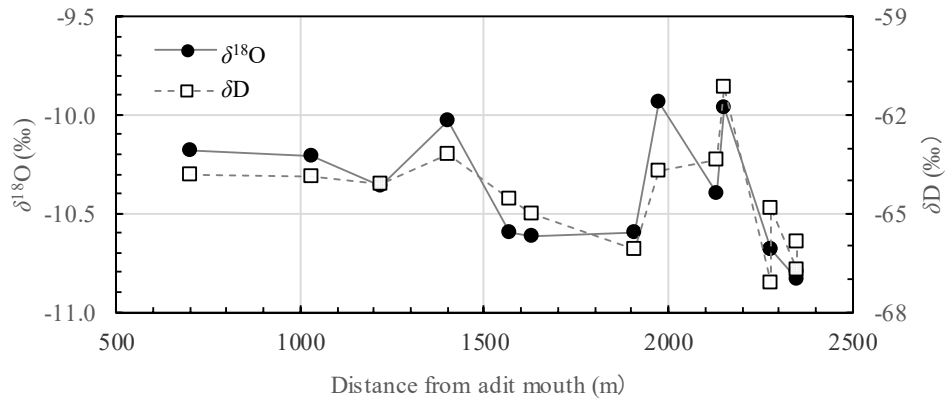


Fig. 7. Spatial distribution of δD and $\delta^{18}\text{O}$ values of mine drainage samples along the drainage level.

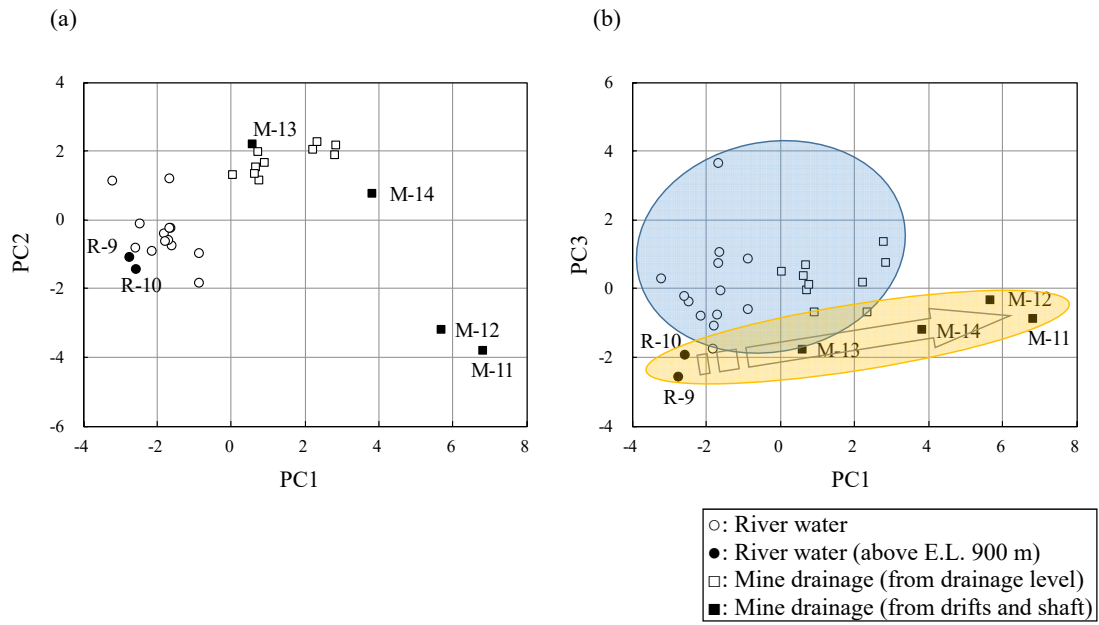


Fig. 8. PCA principal component scores of (a) PC2 vs. PC1 and (b) PC3 vs. PC1.

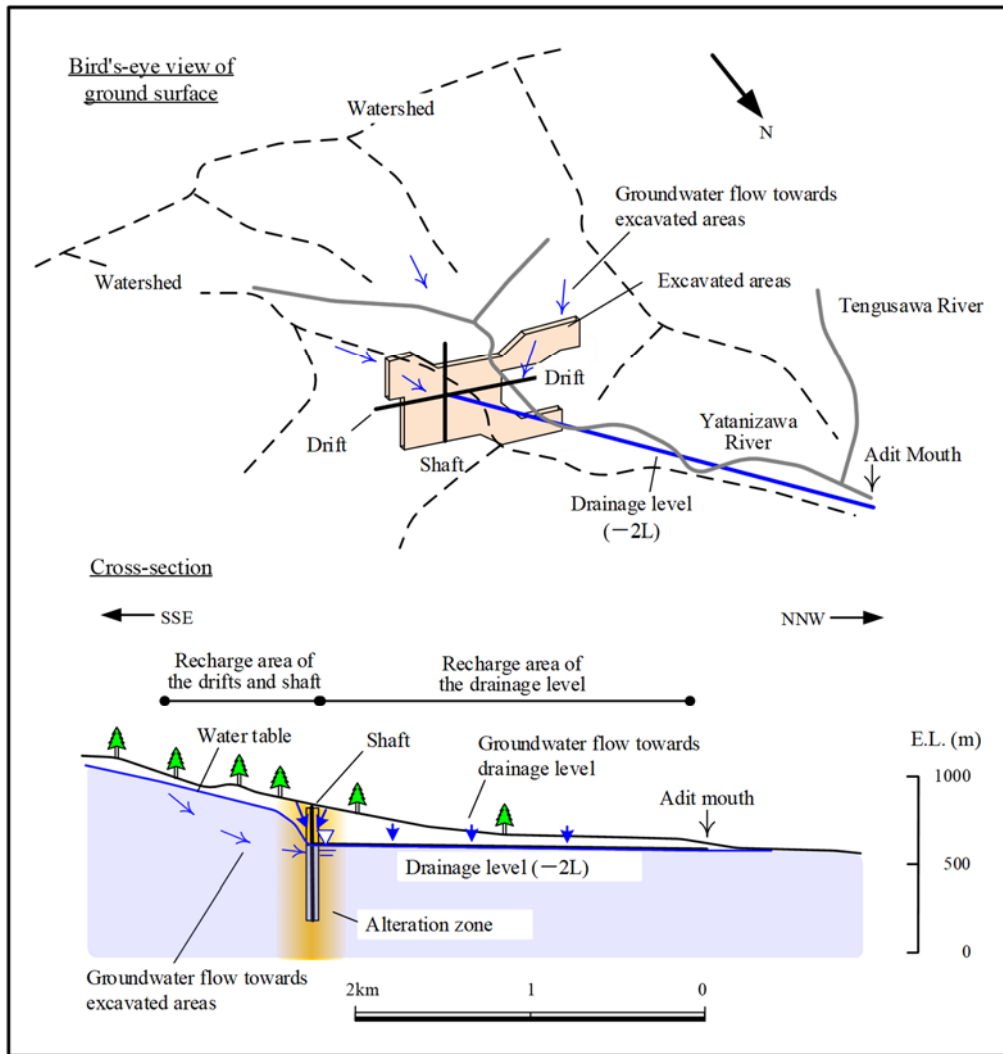


Fig. 9. Conceptual model of groundwater flow and discharge.