

Supplementary Figure 1: Geological maps of the Abitibi Greenstone Belt, Superior Province of the Canadian Shield (inserted panel; modified from Hoffman¹), and the geoloic setting for the Kidd Creek mine, Timmins, Ontario (modified from Barrie²).



Supplementary Figure 2: Simplified stratigraphic column of the rocks bounding the ore deposits at Kidd Creek (modified from Prior et al.³). Formations are shown in original approximately horizontal depositional orientation. Deposits laid down originally approximately horizontally are now steeply dipping sub-vertical units, with boreholes drilled approximately perpendicular to the formation boundaries and intersecting the major units.



Supplementary Figure 3: Simplified stratigraphy of Kidd Creek geologic setting. Massive hydrothermal deposits and other units are shown in original approximately horizontal depositional orientation (after Bleeker et al.⁴). Deposits laid down originally approximately horizontally are now steeply dipping sub-vertical units.



Supplementary Figure 4: Distribution patterns of δ^{34} S and Δ^{33} S for dissolved sulfate in fracture waters and sulfide minerals in the Kidd Creek ore deposit. Most data are from the massive sulfide ores (n = 36; second panel)⁵. Fewer data are available for sulfide sinters (n = 3)⁵ and sulfide turbidites (n = 32)^{5,6} (bottom two panels) which overlap the Δ^{33} S values of the fracture waters and massive sulfides but also show a wider range of values. The mineralization and associated sulfur budget at Kidd Creek is dominated by the massive sulfide ore bodies⁷ and hence these values provide the most representative range for the host rocks.



Supplementary Figure 5: SEM images of sulfide minerals in the Kidd Creek ore samples show various surface morphological features. The mullions and trigonal cavities in panel A

and sharp edges in panel E are likely caused by brittle breakage during either drilling and/or sample preparation for SEM analysis. In contrast, panels B-D (C is an inset of B) show clear evidence of dissolution pits, and the smooth channels on panels F-H (H is an inset of G) are interpreted as etching channels resulting from water-rock interaction. Panels B-D show microcavities with various shapes and sizes on the surfaces of two pyrite grains. The random distribution of these cavities, and their heterogeneous shapes and sizes are inconsistent with fluid inclusions, which typically exhibit a linear or zoned distribution pattern. The cavities are instead very similar to dissolution pits observed in the laboratory experiments of acidic dissolution of pyrite⁸. Panel F shows complex dissolution features on the surface of a chalcopyrite grain that may be associated with etching products by the contacting water. Several stripes at scales of >10 µm extend in different directions and are always associated with abundant (sub)perpendicular distributary lineae (a few to sub µm scale). Some of the stripes have been truncated by larger, more irregular channels. The small lineations may represent a relatively low degree of etching whereas the irregular channels, which are likely produced by coalescent etch pits⁸, represent higher degrees of etching. Panel G shows a few sphalerite grains in a carbonate-rich domain. The grains on the lower half of the image have pits of varying sizes on their top surfaces. Possible reaction cortexes are also seen on all the grains, one of which is enlarged in panel H. The surface of the grain on H clearly shows pronounced roughening and depassivation with crisscross channels resembling etched valleys on the sulfide cortex. The dissolution and etching features described above are commonly seen on the surface of sulfide minerals in rocks from the Kidd Creek mine. Mineral abbreviation: Cc = calcite; Ccp = chalcopyrite; Qz = quartz; Py = pyrite; Sp= sphalerite.



Supplementary Figure 6: SEM images show multiple episodes of re-precipitation of secondary sulfide minerals in primary pyrite mineral grains in the host rocks. (A) Anhedral secondary sulfide minerals, mainly galena (Gn) and sphalerite (Sp), both shown as high brightness in this image because of the high mean atomic mass of PbS and ZnS, completely filled the relatively large angular void spaces (a few to tens of μ m) in a primary pyrite (Py) grain (grey area). The characteristics of the void spaces that host Gn and Sp are distinct from dissolution pits, which are much smaller and smoother (Supplementary Figure 5). The

dissolution pits and etching lineations are widespread on Py surfaces, as well as on the galena and sphalerite fillings. These features indicate that Gn and Sp filling the void spaces were deposited at a relatively early stage, though it is not possible to determine whether they were precipitated in high-temperature hydrothermal fluid or subsequently in lower temperature fracture water. In contrast, some other sulfide fillings, i.e., Pb-coated Py (bright grains) in panel B and "seamount chains" made of Sp (bright grains) in panel C, display a different style of precipitation. Their common features include (1) grains that are mainly composed of small (submicron) euhedral octahedral crystals, (2) without any obvious dissolution features that are instead observed on the surface of surrounding pyrites, and (3) incomplete filling of void spaces. These features suggest at least two conditions. Firstly, secondary sulfide minerals may have precipitated relatively recently and have not yet been influenced much by dissolution. Secondly, the growth rate of the secondary sulfide minerals has been very low, suggesting a dissolved sulfide-poor environment. This latter point suggests a lower temperature system — a hightemperature fluid would likely assimilate a significant amount of sulfide from water-rock reaction. The low-temperature and sulfide-poor conditions are consistent with our model suggesting extremely slow reaction rates of indirect radiolytic oxidation of sulfides and subsequent sulfate reduction during water-rock interaction.

Mineral	Mineral	Metal	Metal	Metal molar	Metal	S Equivalent	Mineral	Mineral	Mineral weight	Mineral
	formula		quantity	mass	quantity	quantity	equivalent	molar mass	$(\times 10^{12} \text{ gram})$	wt%
			(Mt)	$(g mol^{-1})$	$(\times 10^9 \text{ mol})$	(mole)	quantity (mol)			
Sphalerite	ZnS	Zn	8.4¶	65.38	128.48	128.48	128.48	97.44	12.52	17.9
Chalcopyrite	CuFeS ₂	Cu	3.1 [¶]	63.546	48.78	97.57	48.78	183.51	8.95	12.8
Galena	PbS	Pb	0.3 [¶]	207.2	1.45	1.45	1.45	239.26	0.35	0.5
		Total S	33¶	32.06						
Pyrite	FeS ₂		Unknown		Unknown	801.83	400.91	119.965	48.10	68.8

Supplementary Table 1: Mass balance calculation of pyrite concentration in massive sulfide ores.

[¶]Data from Hannington et al.⁹

	^{238}U	^{235}U	²³² Th	^{40}K	Pyrite concentration	
					3.5%	38%
Decay energy (Mev)	44.2	47.1	39.9	0.71		
Concentration $(ppm)^{\uparrow}$	0.90-1.99	0.0066-0.014	4.31-9	1.73-2.34		
Half-life τ (yr)	4.468E+9	7.038E+8	1.405E+10	1.248E+9		
Atomic mass $(g mol^{-1})$	238.05	235.04	232.04	39.96		
D^{app} (10 ⁻³ Gy·yr ⁻¹)	2.51-5.52	0.125-0.275	3.53-7.37	1.65-2.23		
D_{Total}^{app} (10 ⁻³ Gy yr ⁻¹)			7.81-15.40			
$D_{Total}^{abs} \left(10^{-5} Gy \ yr^{-1}\right)^{\uparrow}$			3.81-9.87			
$P_{(M)Sulfate} (10^{-14} \text{ mol } m^{-2} \text{ yr}^{-1})$			8.01-20.72			
$P_{(C)Sulfate} (\mu M yr^{-1})^{\ddagger}$					1.7-4.4 x 10 ⁻⁵	1.9-4.9 x 10 ⁻⁴
$C_{Sulfate}$ (μM) accumulated over the last 2.60 billion years [‡]					4.4-11.4 x 10 ⁴	4.9-12.7 x 10 ⁵
$C_{Sulfate}$ (μM) accumulated over the last 1.1 billion years [‡]					1.9-4.9 x 10 ⁴	2-5.4 x 10 ⁵
Time to produce 97µM sul			2.2-5.6	0.2-0.5		
Time to produce 376µM su			8.4-21.8	0.8-2.0		

Supplementary Table 2: Calculation of sulfate production by indirect radiolytic oxidation of pyrite (IROP).

[†] The range is defined by two groups of concentration data for U, Th and K: 0.91 ppm, 4.31 ppm and 1.48% ¹⁰ to 2 ppm, 9 ppm and $2\%^{11}$ and the isotope concentrations in each element: 99.275% ²³⁸U in U, 0.721% ²³⁵U in U, 100% ²³²Th in Th, and 0.0117% ⁴⁰K in K.

[‡] Calculated by Equation 2 with W value of 0.0043 and S_i value from 1.5 (for α particle) to 1.14 (for γ particle); see text for data sources.

[‡] Considering IROP-derived sulfate only; assuming no sulfate from other sources and not taking into account sulfate loss through sulfate reduction process.

Supplementary Table 3: Calculation	of sulfate loss by thermochemical sulfate reduction
based on a kinetic isotope model.	

Temperature (°C)	50	100
*3	-27.6‰	-19.7‰
$\delta^{34} S_{Initial}$		1.0‰ -5.2‰
$\delta^{34}S_{Final}$		5.3‰ -8.5‰
Remaining sulfate fraction (f)	0.76-1.0	0.68-0.99
Maximum sulfate loss fraction	0.24	0.32
Maximum sulfate Loss/Remaining ratio	0.31	0.46
Sulfate loss in $\mu M^{\$}$	30-116	45-173

*Data from Kiyosu & Krouse¹².

^{\uparrow}Based on the δ^{34} S range of pyrites (-0.5‰ to 1.8‰)⁵ in massive sulfide and the δ^{34} S increase of 1.5‰ to 3.4‰ in sulfate product by IROP¹³.

 $^{\$}$ Calculated from the concentration range (97-376 μ M) of the remaining sulfate in the fracture waters (Main text Table 1).

Supplementary Table 4: Calculation of sulfate loss by thermochemical sulfate reduction based on an equilibrium isotope model.

Temperature (°C)	50	100	
α*	0.942	0.954	
$\delta^{34} S_{\text{Initial}}$	1.	.0‰ -5.2‰	
$\delta^{34}S_{Final}$	5.3‰ -8.5‰		
Remaining sulfate fraction (<i>f</i>)	0.87-1.0	0.84-1.0	
Maximum sulfate loss fraction	0.13	0.16	
Maximum sulfate Loss/Remaining ratio	0.15	0.19	
Sulfate loss in $\mu M^{\$}$	14-55	19-73	

*Calculated from Shanks et al.¹⁴

[†]Based on the δ^{34} S range of pyrites (-0.5‰ to 1.8‰)⁵ in massive sulfide and the δ^{34} S increase of 1.5‰ to 3.4‰ in sulfate product by IROP¹³.

 $^{\$}$ Calculated from the concentration range (97-376 μ M) of the remaining sulfate in the fracture waters (Main text Table 1).

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