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- **1 GSA Data Repository**
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- 3 Supplemental Materials for Zinc isotope evidence for paleoenvironmental changes during
- 4 Cretaceous Oceanic Anoxic Event 2
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- 6 Authors: Xi Chen, Bradley B. Sageman, Hanwei Yao, Sheng'Ao Liu, Kaibo Han, Yi Zou, and
- 7 Chengshan Wang
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16 **1. STUDIED SECTION**

The Gongzha section is located in the Tingri area of Tibet, which tectonically belongs to the Tethys Himalayas (Fig. DR1). Late Cenomanian to Maastrichtian marine strata of pelagic depositional environments are well developed in this section (Wang et al., 2004). During the Cretaceous, the section was located on the northern margin of the Indian continent and the south margin of the Eastern Tethys Ocean. Because a large fraction of sediments deposited in the East Tethys Ocean were subducted beneath the Asian continent, sedimentary records in the Tethys Himalayas provide unique opportunities to study the paleoceanography andpaleoenvironment of the Cretaceous of the East Tethys realm.

25 The Gongzha section yields abundant and poorly to moderately preserved foraminifera, on which biostratigraphy was established (Wan et al., 2003; Li et al., 2006). Li et al. (2006) 26 published a secular $\delta^{13}C_{carb}$ curve for Upper Cretaceous of this section. Based on bio- and 27 chemostratigraphic correlation, the OAE 2 equivalent level is identified with the distinct $\delta^{13}C$ 28 29 positive excursion at the Cenomanian-Turonian transition. The OAE 2 interval, as defined by 30 its characteristic carbon isotope excursion (CIE) in Gongzha is ~ 28 m in thickness. The depositional environment is outer ramp (Wang et al., 2004). During field logging, Li and his 31 32 colleagues painted depth numbers on the rocks every 5 meters. These marked depths acted as 33 a coherent stratigraphic framework for subsequent studies, e.g., Li et al. (2017), Zhang et al. 34 (2019), and this study.

35 Although anoxic to euxinic conditions prevailed during OAE 2 in many basins, evidence 36 does not suggest the same redox conditions characterized the south margin of the eastern 37 Tethys Ocean. Elemental geochemistry, including redox indicators such as V and Cr, indicate 38 that the water column was oxygenated during the event (Bomou et al., 2013). In contrast to these observations, recent $\delta^{15}N$ data were interpreted to reflect expansion of an anoxic 39 environment during the C-T interval (Zhang et al., 2019), however, the $\delta^{15}N$ record from 40 41 Gongzha does not show the significant negative excursions observed in other OAE2 sites and 42 interpreted to reflect nutrient and redox dynamics (Kuypers et al., 2004; Junium and Arthur, 43 2007).

Age constraints for OAE 2 in this section are based on planktonic foraminifera and an

45	astronomical time scale (ATS). The lower 70 m of the section contains R. cushmani, W.
46	archaeocretacea, and H. helvetica biozones (Wan et al., 2003; Li et al., 2006; Bomou et al.,
47	2013). The astronomical time scale from the lower part of the Gongzha section was developed
48	by Li et al. (2017). Based on the ATS, Li et al. (2017) estimated that the duration of the $\delta^{13}C$
49	excursion (defined by return to background δ^{13} C values) is ~870 kyr (this is the longer of two
50	common definitions of the event discussed in the literature: e.g., Sageman et al., 2006). They
51	divided the δ^{13} C curve across OAE 2 into 6 segments, including pre- and post-excursion units.
52	This study largely followed the scheme of Li et al. (2017) to divide the segments of $\delta^{13}C$
53	curve across OAE 2, except for interpretation of the initial onset of the OAE 2 carbon isotope
54	excursion. There is a fairly consistent background $\delta^{13}C_{\text{carb}}$ signal with relatively minor
55	variance in strata prior to the OAE 2 positive excursion (from the base to \sim 40 m), defined as
56	segments C1 and C2. Although initiation of the OAE 2 CIE was placed at ~37 m by Li et al.
57	(2017), δ^{13} C values are similar to background values (< 2‰) and start to increase rapidly
58	from ~40 m, which is recorded in the $\delta^{13}C$ dataset of both previous publications (Li et al.,
59	2017) and this study (Fig. 2). Therefore, we suggest that the OAE 2 CIE initiates at \sim 40 m in
60	the section. Segment C3 (from \sim 40 m to \sim 48 m) is characterized by rapid increase in values of
61	$\delta^{13}C_{carb}$, marking the onset of OAE 2 and lasting for ~280 kyrs. In segment C4 (from ~48 m to
62	64 m; ~350 kyr), the $\delta^{13}C_{carb}$ values persist at ~3‰ followed by stepwise recovery of $\delta^{13}C_{carb}$
63	values to pre-excursion levels (segment C5). The ATS (Li et al., 2017) provides a foundation
64	for future study on the evolution of paleoenvironment during OAE 2.

66 2. ANALYTICAL METHODS

67 2.1 Procedure for Zn isotope measurement

68	A leaching procedure used to prepare for Zn isotope measurements strictly followed the
69	method of Liu et al. (2017). Samples were carefully checked to avoid visible veins and
70	fractures, and were ground to ~ 200 mesh with an agate mortar that was carefully cleaned
71	with Milli-Q water (18.2 ${\rm M}\Omega$). The leaching steps were performed in a clean room under
72	laminar flow hood (class 100) to minimize Zn contamination. Hydrosoluble salts and
73	exchangeable fraction on clays were removed before leaching of carbonates using Milli-Q
74	water and 1 N ammonium acetate (NH ₄ AC), alternately. The carbonate fraction was
75	selectively dissolved with two steps of 0.05 M super-pure acetic acid in a thermostat at 65 $^\circ\!\mathrm{C}$,
76	until no bubbles were generated. The supernatants were collected after 20 min of
77	centrifugation, followed by filtration. The leached supernatants were dried at 80 $^\circ\!\!\mathrm{C}$ and
78	added with 2 ml of 8 M HCl. This step was repeated twice to remove acetic acid. Finally, the
79	solution was prepared in 1 ml of 8 M HCl for chemical anion-exchange separation.
80	The leaching solutions were dried at 80 °C. Zinc was purified by an ion-exchange
81	chromatography using Bio-Rad strong anion resin AG-MP-1M. 2 ml pre-cleaned resin was
82	loaded onto the column. Matrix elements were eluted in the first 10 ml of 8 N HCl. Copper
83	and iron were then collected in the following 24 ml of 8 N HCl +0.001% $\rm H_2O_2$ and 18 ml of
84	2N HCl. Zinc was collected in the subsequent 15 ml of 0.5N HNO ₃ . This procedure allows Ca
85	to be completely separated from Zn and a 100% recovery for Zn. The Zn fractions were

repeatedly dried and dissolved with 3% HNO₃ to remove all chlorine prior to isotope analysis.
Zinc isotopic ratios were measured using a Neptune plus multi-collector inductively

coupled plasma mass-spectrometry (MC-ICP-MS) at the Isotope Geochemistry Laboratory of

89	the China University of Geosciences (Beijing). Sample-standard bracketing method was used
90	to correct for instrumental mass fractionation. The samples and standards are run in ~ 200 ppb
91	diluted solution dissolved in 3% (v) HNO ₃ . The take-up time is 80 s. The Zn isotope data for
92	leaching solutions and residues are reported in Table DR. 2. Zn isotope data of the leachates
93	are the mean of the two steps of leaching on the samples, which agree with each other within
94	$\pm 0.1\%$. All samples analyzed in this study yield a slope of 2.08 in a δ^{68} Zn– δ^{66} Zn cross-plot,
95	which is consistent with the mass-dependent line with a slope of 2, indicating mass-dependent
96	Zn isotope fractionation and no analytical artifacts from unresolved isobaric interferences on
97	Zn isotopes.

99 2.2 Measurements of major and trace element concentrations of bulk sample

100 Analyses of major element concentrations were conducted at the State Key Laboratory 101 of Geological Processes and Mineral Resources, China University of Geosciences, Beijing and the Wuhan SampleSolution Analytical Technology Co., Ltd., Wuhan, China. For 102 103 elemental oxide concentration analysis, a 0.7g sample (200 mesh) was mixed with 7.0g of 104 lithium borate and fused to glass beads. The addition of a small amount (100 mg) of the 105 halide LiBr acted as a release agent when using platinum molds. The mixture was heated in a 106 high frequency fusion machine at ~1000 °C, until thoroughly melted. Then the fused beads 107 were loaded into the XRF instrument for testing on a Shimadzu wavelength dispersive X-ray 108 fluorescence spectrometer. Concentrations of the major elements are listed in Table DR. 1. 109 Trace element analysis of whole rock samples were conducted on an Agilent 7700e

110 ICP-MS at the Wuhan SampleSolution Analytical Technology Co., Ltd., Wuhan, China. The

111	detailed sample-digesting procedure was as follows: (1) Sample powders (200 mesh) were
112	placed in an oven at 105 for drying of 12 hours; (2) 50 mg sample powder was accurately
113	weighed and placed in a Teflon bomb; (3) 1 ml HNO ₃ and 1 ml HF were slowly added into
114	the Teflon bomb; (4) The Teflon bomb was put in a stainless steel pressure jacket and heated
115	to 190 in an oven for >24 hours; (5) After cooling, the Teflon bomb was opened and placed
116	on a hotplate at 140 and evaporated to incipient dryness, and then 1 ml HNO ₃ was added
117	and evaporated to dryness again; (6) 1 ml of HNO ₃ , 1 ml of Milli-Q water and 1 ml internal
118	standard solution of 1ppm In were added, and the Teflon bomb was resealed and placed in the
119	oven at 190 for >12 hours; (7) The final solution was transferred to a polyethylene bottle
120	and diluted to 100 g by the addition of 2% HNO ₃ . V/Cr ratios of bulk sample (Table DR. 2)
121	were employed to investigate the redox condition of the studied interval (Fig. 2).

123 2.3 Measurement of trace element concentrations in leachates

124 Sample preparation for measurement of trace element concentrations were largely similar 125 to the method described above for the Zn-isotope analysis, with a few differences: 1) 126 exclusion of treatment of the Milli-Q water and 1 N ammonium acetate to remove the 127 exchangeable fraction; 2) The leached supernatants were dried at 80 and added with 0.5 ml 128 of HNO3; 3) 5 ml Mill-Q water and 0.3 ml of 6% HNO3 were added into the beaker at 80 129 for 2 hours to dissolve the sample. The final solution was transferred to a polyethylene bottle 130 and diluted to 100 g by the addition of 2% HNO₃. Element abundances were analyzed on an 131 Agilent 7700e ICP-MS at the Wuhan SampleSolution Analytical Technology Co., Ltd., Wuhan, China. The Zn concentration in carbonates is calculated from Zn contents in the 132

leaching solutions divided by the contents of carbonates (Zn/Ca, μ g/g), the latter of which are

134 calculated based on major elements in the bulk rocks.

Because Sr/Ca and δ^{18} O values may be altered significantly in cases of major change in carbonate precipitation rate (Lorens, 1981; Watkins et al., 2014), or during extensive carbonate diagenesis (Swart, 2015), the fact that there is insignificant co-variation between δ^{66} Zn values and these two proxies (Figs. DR2 and DR3) indicates that neither process is responsible for the observed excursion in δ^{66} Zn.

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141 **3. Figures**

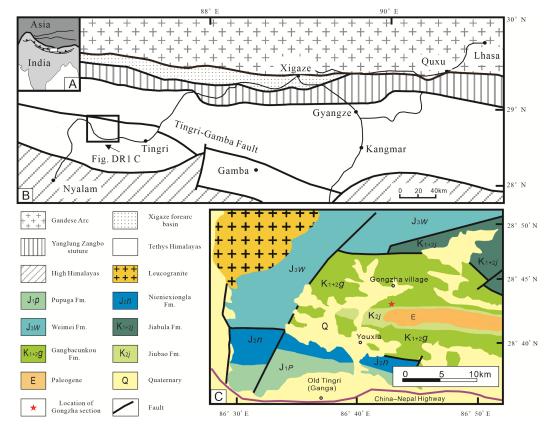


Fig. DR 1 Tectonic units of the southern Tibet (A and B) (Li et al., 2006) and geological map
of the studied area showing the location of studied section (C) (modified from Hu et al.,
2010).

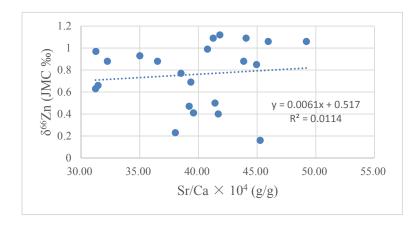
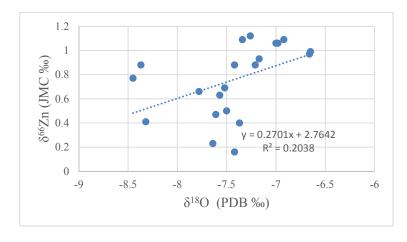


Fig. DR 2 Cross plot of δ^{66} Zn–Sr/Ca of carbonates showing the absence of co-variation. The lack of δ^{66} Zn– δ^{18} O (Fig. DR 2) and δ^{66} Zn–Sr/Ca covariance indicates that the Zn isotope shift

149 is not related to precipitation rate or diagenesis.

150





153 Fig. DR 3 Cross plot of δ^{66} Zn– δ^{18} O of carbonates showing the absence of co-variation. The

154 lack of δ^{66} Zn– δ^{18} O and δ^{66} Zn–Sr/Ca (Fig. 3 in main text) covariance indicates that the Zn

isotope shift is not related to precipitation rate or diagenesis.

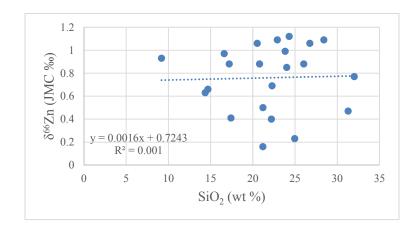




Fig. DR 4 Cross plot of δ^{66} Zn vs. SiO₂ of bulk sediments from the Gongzha section. The absence of correlation suggests that the temporal δ^{66} Zn variations are not caused by adsorption of Zn onto silicate or clay phases.

162

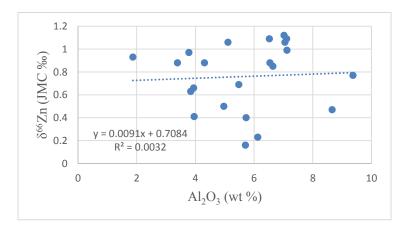
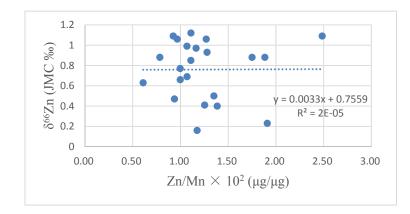




Fig. DR 5 Cross plot of δ^{66} Zn vs. Al₂O₃ of bulk sediments from the Gongzha section. The absence of correlation suggests that the temporal δ^{66} Zn variations are not caused by

adsorption of Zn onto silicate or clay phases.





169 Fig. DR 6 Cross plot of δ^{66} Zn vs. Zn/Mn in leachates from the Gongzha section. The 170 absence of correlation suggests that the temporal δ^{66} Zn variations are not caused by leaching 171 of ferromanganese coatings.

173 **4. Tables**

Sample	Height/m	SiO ₂	TiO ₂	Al_2O_3	TFe ₂ O ₃	MnO	MgO	CaO	K ₂ O	Na ₂ O	P_2O_5	LOI	Total
Number	incigit/iii	5102	1102	11203	11 0203	WIIIO	MgO	CaO	K ₂ O	11420	1 205	LOI	Total
09GZ240	27.19	20.52	0.6	5.11	3.67	0.09	1.04	37.26	0.7	0.27	0.09	30.49	99.84
09GZ261	29.21	28.41	0.93	7.11	5.36	0.09	1.23	29.08	1.29	0.3	0.07	25.39	99.26
09GZ274	30.56	26.03	0.83	6.54	5.23	0.1	1.24	30.98	1.21	0.21	0.08	26.62	99.07
09GZ292	32.21	24.3	0.81	7.02	4.26	0.09	1.18	32.88	1.18	0.31	0.1	27.82	99.95
09GZ302	33.25	26.75	0.88	7.05	5.43	0.09	1.32	30.4	1.32	0.22	0.1	26.09	99.65
09GZ333	36.1	23.83	0.69	7.12	5.22	0.11	1.24	32.49	1.01	0.13	0.25	27.56	99.65
09GZ349	37.49	24.02	0.76	6.64	6	0.09	1.38	31.61	1.06	0.19	0.12	27.28	99.15
09GZ367	38.92	22.9	0.76	6.52	4.22	0.06	1.21	34.23	1.18	0.71	0.08	28.41	100.28
TOC1	40.05	32.01	1.08	9.37	5.99	0.04	1.22	24.66	1.92	0.37	0.11	22.62	99.39
TOC3	41	31.29	1.08	8.66	5.79	0.05	1.21	26.7	1.46	0.33	0.09	23.31	99.97
TOC6	42.05	22.28	0.67	5.48	4.85	0.06	1.11	34.94	0.74	0.22	0.6	28.39	99.34
TOC8	42.8	21.20	0.65	4.97	3.19	0.05	0.83	36.71	0.82	0.49	0.11	30.66	99.69
TOC-11	43.9	17.41	0.47	3.96	2.43	0.04	0.65	40.22	0.60	0.57	0.08	33.28	99.71
TOC-14	45.1	21.19	0.62	5.71	5.27	0.04	1.00	34.55	0.81	0.30	0.07	29.97	99.53
TOC-25	49.75	22.19	0.77	5.73	3.13	0.03	0.87	35.43	1.06	0.45	0.07	30.06	99.77
TOC-33	52.6	24.96	0.82	6.13	3.98	0.03	1.13	33.16	1.03	0.36	0.07	28.50	100.15
TOC36	53.75	17.21	0.54	3.39	2.17	0.03	0.86	41.66	0.55	0.13	0.06	33.05	99.65
TOC43	56.5	9.18	0.23	1.87	1.77	0.04	0.71	48.38	0.23	0.18	0.05	37.39	100.03
TOC54	61.8	14.36	0.36	3.84	2.68	0.04	0.95	43.27	0.34	0.14	0.09	34.16	100.23
TOC56	63.1	14.68	0.43	3.94	2.41	0.04	0.95	42.91	0.43	0.1	0.05	34.21	100.15
TOC59	64.6	20.8	0.63	4.31	2.83	0.03	0.87	39.07	0.71	0.2	0.06	31.21	100.72
09GZ654	66.14	16.61	0.4	3.78	2.74	0.06	0.97	40.96	0.43	0.1	0.06	33.21	99.32

 175
 Table DR 1 Weight concentration of elemental oxides (wt %)

177 Table DR 2 Zinc isotope ratios of leached carbonate fraction, δ^{18} O values (‰), V/Cr and Sr/Ca ratios of bulk sample, Zn concentration and Zn/Mn and

178 Zn/Ca ratios in leachates. Sr/Ca ratio values are multiplied by 10^4 . The concentration of Ca is calculated from the concentration of CaO: Ca (%) = CaO (%)

Height/m	δ^{66} Zn (‰)	2SD	δ^{68} Zn (‰)	2SD	δ^{68} Zn / δ^{66} Zn	δ ¹⁸ O (‰)	V/Cr (µg/µg)	Sr (µg/g)	Sr/Ca	Zn (µg/g)	Zn/Mn	Zn/Ca
27.19	1.06	0.04	2.13	0.01	2.00	-6.98	1.24	1309	51.63	6.50	0.97	24.42
29.21	1.09	0.01	2.16	0.01	1.99	-7.34	1.15	915	44.20	6.12	0.93	29.47
30.56	0.88	0.05	1.81	0.03	2.06	-7.21	1.23	970	44.28	5.94	0.78	26.84
32.21	1.12	0.02	2.25	0.05	2.01	-7.26	1.26	982	42.28	7.24	1.11	30.85
33.25	1.06	0.01	2.13	0.04	2.02	-7.00	1.22	998	48.59	8.69	1.27	40.03
36.10	0.99	0.00	1.99	0.02	2.01	-6.65	1.31	946	40.93	8.25	1.07	35.54
37.49	0.85	0.02	1.80	0.02	2.12	N/A	1.25	1015	48.05	6.53	1.11	28.94
38.92	1.09	0.03	2.19	0.01	2.02	-6.92	1.25	1009	43.42	10.6	2.49	43.26
40.05	0.77	0.02	1.70	0.06	2.21	-8.45	1.16	678	38.43	2.97	1.00	16.84
41.00	0.47	0.06	4.85	0.06	2.00	-7.61	1.07	748	39.21	3.21	0.94	16.82
42.05	0.69	0.06	0.92	0.06	1.97	-7.52	1.46	982	39.32	4.58	1.07	18.37
42.80	0.50	0.06	1.35	0.01	1.97	-7.50	1.31	1086	41.41	5.38	1.35	20.52
43.90	0.41	0.06	1.01	0.06	2.01	-8.32	1.54	1137	45.83	3.65	1.25	12.71
45.10	0.16	0.06	0.83	0.06	2.03	-7.42	1.35	1117	45.26	3.20	1.18	12.98
49.75	0.40	0.06	2.12	0.06	2.07	-7.37	1.10	1055	41.71	3.03	1.39	11.96
52.60	0.23	0.09	0.81	0.06	1.99	-7.64	1.11	901	38.03	4.56	1.91	19.27
53.75	0.88	0.04	0.45	0.08	1.98	-7.42	1.00	1087	39.09	4.04	1.75	13.58

179 \times 40/56. Zn/Mn and Zn/Ca ratio values are multiplied by 10².

56.50	0.93	0.03	1.88	0.07	2.03	-7.17	0.95	1210	36.76	3.74	1.28	10.82
61.80	0.63	0.02	1.89	0.06	3.01	-7.57	1.04	966	31.23	1.87	0.61	6.05
63.10	0.66	0.00	1.37	0.03	2.08	-7.78	1.09	964	31.51	2.72	1.00	8.87
64.60	0.88	0.05	1.78	0.01	2.03	-8.37	1.04	900	33.05	4.31	1.88	15.44
66.14	0.97	0.05	2.04	0.03	2.11	-6.66	1.10	915	31.56	4.93	1.16	16.84

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