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Measurements of seismic waves induced by highvelocity impacts: Implications for seismic shaking surrounding impact craters on asteroids

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- 18 **Proposed running head:** Impact-induced seismic waves at high-velocity impacts
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28 Abstract

On asteroids, impact-induced seismic shaking is an important geological process related 29 30 to the modifications of the surface features that are due to the movement of regolith grains 31 on the asteroid surface. Mass movements caused by seismic shaking (e.g., landslides and 32 crater erasures) were recently observed by several spacecrafts. To elucidate the seismic 33 shaking areas induced by impacts on asteroids, it is necessary to determine the excitation 34 and the decay processes of impact-induced seismic waves. Here we conducted impact 35 cratering experiments using 500-µm quartz sand to simulate an asteroid surface at the 36 Institute of Space and Astronautical Science (ISAS), Japan Aerospace Exploration 37 Agency (JAXA). We used 4.75-mm- and 2-mm-dia. projectiles with density from 14.9 to 1.2 g cm⁻³; the impact velocity ranged from 7 to 0.2 km s⁻¹. We measured the acceleration 38 39 waveform by using three to five accelerometers setting on the target surface, and we 40 examined three parameters characterizing the impact-induced seismic wave with varying distances from the impact point: the propagation velocity (V_{prop}) , the maximum 41 42 acceleration (g_{max}) , and the half period of the first upward peak (T_{half}) . The V_{prop} was obtained to be 52.4 \pm 7.2 m s⁻¹, regardless of the impact velocity and the projectile 43 properties, and the T_{half} slightly depended on the impact velocity as $v_i^{0.14}$ but was 44 45 independent of the projectile properties. The g_{max} had a close relationship with the distance from the impact point normalized by the crater rim radius, x/R_{rim} , regardless 46 47 of the impact velocity and the projectile properties. The obtained empirical equation was

shown as $g_{\text{max}} = 10^{2.21} (x/R_{\text{rim}})^{-3.18}$. By using the empirical equations we obtained herein, we observed that the seismic energy normalized by the kinetic energy of a projectile, E_s/E_k , decreased with the increase of the normalized distance, x/R_{rim} . We also estimated the area of impact-induced seismic shaking on an idealized small (500 m in diameter) body at the impact velocity of 5 km s⁻¹.

53 Keywords: Asteroids; Asteroids, surfaces; Experimental techniques; Cratering; Impact

54 processes

55 1. Introduction

56 1.1. Seismic shaking on an asteroid surface

Seismic shaking induced by a high-velocity impact is one of the most important 57 58 surface geologic processes that occurs on an asteroid surface. Such shaking frequently 59 modifies the asteroid surface topography. Several planetary explorations recently revealed the detailed surface geology of asteroids, and evidence of regolith migration was 60 identified on the asteroid surfaces. For example, NEAR-Shoemaker and Hayabusa 61 62 spacecrafts obtained numerous close-up images of the asteroids 433 Eros and 25143 63 Itokawa. The images of the asteroid surfaces showed unconsolidated gravel and pebbles which were typically stacked upon each other [Miyamoto et al., 2007]. Regolith migration 64 65 was then identified on the asteroid surface as alignments of boulders, landslides, and 66 crater erasures [Veverka et al., 2001; Cheng et al., 2002; Thomas et al., 2002; Robinson 67 et al., 2002; Miyamoto et al., 2007].

These features could be explained by global-scale and/or local-scale seismic shaking caused by high-velocity impacts on the regolith surface. Impact-induced seismic shaking on a global scale could trigger regolith migration under a vacuum and in the microgravitational environment that is typically present on small asteroids [Miyamoto et al., 2007]. These phenomena could preferentially remove small undulations such as small craters.

74

Asteroid 433 Eros, one of the most intensively studied small extraterrestrial bodies,

75 was revealed to have surface areas at which the crater number density differed 76 considerably. It also appeared that the number density of small impact craters (<~100 m 77 in diameter) was much smaller than that extrapolated from the >100-m impact craters, 78 based on the cumulative size distribution of impact craters detected on the asteroid 79 [Robinson et al., 2002].

80 To study this deficiency in the number of small craters, Richardson et al. [2005] developed a theoretical model for the crater erasure process caused by global seismic 81 82 shaking. They stated that their calculations indicated that the vibration of the regolith 83 layer covering the entire asteroid surface was responsible for erasing the craters. A 84 deficiency in the number of craters with diameters <500 m was observed around 85 Shoemaker crater (within 9 km from the center of the crater), and this unique feature of 86 low crater density suggested that the crater degradation could have been caused by local 87 seismic shaking induced by the formation of Shoemaker crater [Robinson et al., 2002].

88 Several research groups have carried out numerical simulations to investigate the 89 effect of impact-induced seismic shaking on the motion of particles in the regolith layer 90 of an asteroid surface. Asphaug and Melosh [1993] performed a numerical simulation 91 using a hydrodynamic code for a high-velocity impact, and they examined the seismic 92 resurfacing of the asteroid surface. Greenberg et al. [1994, 1996] proposed a 'global jolt' 93 resulting from large impacts on the surfaces of an asteroid such as 243 Ida and 951 Gaspra, 94 and they discussed the effect of the jolt on the crater size distributions. The results of both

6

95 the Asphaug and Melosh and Greenberg et al. studies suggested that there is a key 96 parameter that induces the resurfacing that is related to impact craters on asteroids, that 97 is, an impact-induced seismic efficiency factor that could be defined by the ratio of the 98 seismic energy to the kinetic energy of the impactor.

99 Such an impact-induced seismic efficiency factor is poorly understood, and only a 100 few impact experiments have been conducted to evaluate impact-induced seismic waves 101 in a laboratory setting. McGarr et al. [1969] conducted high-velocity impact experiments 102 using loose sand and epoxy-bonded sand as the targets and a Lexan projectile at impact velocities of 7–0.8 km s⁻¹. They observed the impact-induced seismic waves by setting 103 104 accelerometers on the target surface. Their results demonstrated an impact-induced seismic efficiency factor of 6.0×10^{-6} for the loose sand and 7.6×10^{-5} for the bonded 105 106 sand target.

107 Yasui et al. [2015] conducted impact experiments on targets made of 200-µm-dia. glass beads impacted at the impact velocities of $<150 \text{ m s}^{-1}$ to study the attenuation rate 108 109 of the impact-induced seismic wave. They observed that the maximum acceleration, i.e., $g_{\rm max}$, of the seismic wave decayed with the distance from the impact point normalized 110 by the crater radius, x/R, according to the following power law equation: $g_{\text{max}} =$ 111 $g_0(x/R)^{-2.2}$ within x/R = 5. Their calculations revealed that the average impact-112 induced seismic efficiency factor was 5.7×10^{-5} . However, this result was obtained at 113 114 impact velocities $<150 \text{ m s}^{-1}$, which is much slower than the relative collisional velocities

among asteroids. Moreover, spherical glass beads were used as the target, and these beads might be an idealized granular material simulating asteroid regolith. The seismic efficiency factor had a wide range $(10^{-2}-10^{-6})$ depending on the impactor and target properties [Gault and Heitowit, 1963; McGarr et al., 1969; Collins et al., 2005] and the impact energies [Shishkin, 2007].

More realistic numerical simulations were recently conducted by using iSALE to reproduce an impact crater formed on cohesive materials such as porous basalt, quartzite, and sandstone. The estimated seismic efficiency factor ranged from 10^{-3} to 10^{-5} for the elastic waves that were induced by the impact at a site far from the impact point [Güldemeister and Wünnemann, 2017; Wójcicka et al., 2019].

We conducted the present study to determine the attenuation rate of a seismic wave 125 126 propagating through a target simulating an asteroid regolith, and we investigated the 127 impact-induced seismic efficiency factor at high impact velocities (i.e., those comparable 128 to the relative collisional velocities among asteroids). We then carried out impact 129 cratering experiments and measured the impact-induced seismic waves by using 130 accelerometers set on a granular target, following the methods described by McGarr et al. 131 [1996] and Yasui et al. [2015]. For our investigation of the effects of the impact velocity 132 and the projectile properties on the impact-induced seismic waves, the target material 133 (quartz sand) was fixed. We also estimated the impact-induced seismic energy by using a 134 revised sinusoidal wave model delineated by Yasui et al. [2015], and we discuss the impact-induced seismic efficiency factor. We also assessed the seismic shaking area on
an idealized small body by using the obtained empirical equations obtained herein and
the crater scaling laws.

138

139 *1.2.* π -scaling theory

140 Crater scaling theories are essential to evaluations of planetary-scale impact 141 phenomena under various impact conditions. These theories have been tested in 142 laboratory experiments. In the present study, we applied the π -scaling theory proposed 143 by Holsapple [1993] to our experimental results. This conventional theory for cratering processes can be used to extrapolate our laboratory-scale results to planetary-scale 144 145 impacts. We used the following non-dimensional scaling parameters, i.e., π_R , which is 146 related to the crater radius (R_c), and π_2 and π_4 , which are related to various impact 147 conditions:

148
$$\pi_{\rm R} = R_{\rm c} \left(\frac{\rho_{\rm t}}{m_{\rm p}}\right)^{1/3}$$
, (1)

149
$$\pi_2 = \frac{gr_p}{v_i^2},\tag{2}$$

150
$$\pi_4 = \frac{\rho_t}{\rho_p},\tag{3}$$

151 where ρ_t is the bulk density of target, m_p is the projectile mass, ρ_p is the projectile 152 density, r_p is the projectile radius (they are written as $m_p = 4\pi \rho_p r_p^3/3$ for a spherical 153 projectile), g is the gravitational acceleration, and v_i is the impact velocity.

In a gravity regime in which the crater formation process is controlled by the gravity of the body, the following relationship is proposed on the basis of a point source approximation and a coupling parameter, $C = r_p v_i^{\mu} \rho_p^{\nu}$:

157
$$\pi_{\rm R} = H_1 \, \pi_2^{-\frac{\mu}{2+\mu}} \pi_4^{\frac{2+\mu-6\nu}{3(2+\mu)}}, \tag{4}$$

158 where H_1 , μ , and ν are the constants that depend on the target properties, We were able 159 to determine these values in laboratory experiments.

160

161 **2. Experimental methods**

162 2.1. Impact experiments

163 We used quartz sand with a mean grain diameter of 500 µm as a target simulating the 164 regolith layer on an asteroid surface. The grains of the quartz sand have irregular shapes 165 but rounded corners. The measured angle of repose was $33 \pm 2^{\circ}$, which is much higher 166 than that of the 200- μ m spherical glass beads (23 ± 2°) used by Yasui et al. [2015]. The grain density of the quartz sand was 2.7 g cm⁻³ and the calculated bulk density of the 167 168 quartz-sand target was 1.46 g cm⁻³; thus, the estimated bulk porosity was approximately 169 42%. The quartz sand was poured into two stainless-steel containers: the larger 60-cm-170 diameter, 20-cm-deep container was used for the impact velocities >1.1 km s⁻¹, and the smaller 30-cm-diameter, 10-cm-deep container was used for the impact velocities <230 171

172 m s⁻¹. Before every shots, the surface of the target was flattened with the use of a hard 173 metal ruler.



Each of the projectiles was a sphere, made of the following: polycarbonate (PC, density of 1.2 g cm⁻³), soda-lime glass (Gl, 2.5 g cm⁻³), aluminum (Al, 2.7 g cm⁻³), titanium (Ti, 4.6 g cm⁻³), zirconia (ZrO₂, 5.7 g cm⁻³), stainless steel (SUS, 7.9 g cm⁻³), copper (Cu, 9.0 g cm⁻³), and tungsten carbide (WC, 14.9 g cm⁻³). The diameter of the PC projectile was 4.75 mm, and the diameter of the other seven projectiles was 2.0 mm. Each projectile was launched in the direction parallel to the gravity acceleration of the Earth, so that it impacted at a direction perpendicular to the target surface.

To examine the effects of the grain shape on the cratering process, we also conducted low-velocity cratering experiments using the quartz sand at the impact velocity of 170-227 m s⁻¹ in order to compare our results with those of the Yasui et al. [2015] study. It is

difficult to achieve an impact velocity below 1.0 km s^{-1} by using the two-stage gas gun at ISAS, and thus our low-velocity cratering experiments were carried out using the same experimental set up as that used by Yasui et al. [2015].

195

196 2.2. Observations of impact phenomena

197 We used an uniaxial piezoelectric accelerometer (#SV1113, Nippon Avionics, Tokyo) with a charge amplifier (#AG2101, Nippon Avionics) to measure the acceleration 198 199 activated on the target surface affected by an impact-induced seismic wave. The charge 200 sensibility was 5.47 pC s² m⁻¹, and the response frequency of the acceleration was 0.5 201 Hz-10 kHz. For our investigation of the decay process of impact-induced seismic waves, 202 we set three, four, or five accelerometers (Fig. 1b) on the target surface at 35.3 to 5.1 cm 203 from the impact point, outside of the impact crater rim. The measurement direction of the 204 accelerometers were perpendicular to the target surface; the accelerometers could 205 therefore detect the normal components of the acceleration activated on the surface. The 206 accelerometers were buried at the depth of 2.5 cm (Fig. 1c).

Each seismic wave measured by the accelerometers was recorded by a data logger with an A/D conversion rate of 100 kHz (midi LOGGER GL900, Graphtec, Yokohama, Japan). We repeated the cratering experiments using the same impact conditions several times to acquire the seismic wave data at different propagation distances, ensuring that we would have a sufficient amount of data to determine the attenuation rate of the seismic 212 waves.

It is necessary to remove any artificial noise added to the actual signals, because such noises are naturally induced by the instantaneous force generated by the projectile launching. This force could activate the vibration of the target container set in the target chamber. The bottom of the target container was thus insulated by a rubber sheet to remove the activated vibration (Fig. 1a) so that the acceleration of each impact-induced seismic wave was measured precisely.

We observed the impact phenomena by using a high-speed digital camera (Hyper Vision HPV-X, Shimadzu, Kyoto, Japan) to determine the impact time of each projectile on the target surface. The frame rate was set at 10^5 frames s⁻¹. The impact time could be determined within 10 µs. To synchronize the trigger of the high-speed digital camera with the trigger of the data logger, we generated a trigger signal by using a function generator to send the signal to both devices at the moment when the velocity measurement system detected a signal showing the projectile passing through.

After the shot, a photograph was taken just above the target surface for the analysis of the precise distance of each accelerometer from the impact point. We used a digital caliper to measure the rim-to-rim distances in two directions where they were crossed at right angles. The average distance was defined as the rim diameter, D_{rim} . In several shots, we measured a cross-section of the impact crater by using a laser profiler (#LJ–V7300, Keyence, Osaka, Japan) with 10-µm resolution to determine the ratio of the depth to the diameter of the impact crater.

233

235 3.1. Morphology of impact craters

The experimental conditions and the results on the morphology of the impact craters are summarized in Table 1. Fig. 2 shows the crater profiles obtained under various impact conditions. As shown in the figure, all of the impact craters formed in these experiments were categorized as a simple crater with a bowl-shaped depression surrounded by a distinct rim. A bright area was spread over the center of the crater floor in each case; this area might have been comminuted quartz sand grains.

242 The impact crater formed by the PC projectile is shown in Fig. 2a, b. The crater size at $v_i = 6.9 \text{ km s}^{-1}$ was about twice that obtained at 1.1 km s⁻¹. When an impact velocity 243 244 was high (Fig. 2b), several small pieces of black debris were observed on the bright area 245 of the crater floor; these might have been charred broken projectile fragments. Impact 246 craters formed by an Al projectile and a Cu projectile are shown in Fig. 2c, d, and the 247 impact velocity was almost the same: approximately 4 km s⁻¹. The crater size increased 248 with the increase of the projectile density: the crater made by the Cu projectile was 1.3 249 times larger than that by the Al projectile at the density ratio of Cu to Al of approximately 250 3.

Fig. 3a is crater profiles obtained with the PC projectiles at different impact velocities.

Table 1

252 Both the diameter and the depth of the craters clearly became larger as the impact velocity 253 increased. To study the similarity of the crater shapes formed at different impact velocities, 254 we normalized both the vertical and horizontal axes in Fig. 3a by each crater rim radius, $R_{\rm rim}$, where $R_{\rm rim} = D_{\rm rim}/2$, and then we plotted the normalized crater profile at each 255 256 impact velocity (Fig. 3b). The normalized crater profiles were very similar, regardless of 257 the impact velocity. Thus, the similarity of the crater shapes was completely confirmed. Fig. 3c shows the crater profiles obtained at the impact velocity of $v_1 = 4.0 \text{ km s}^{-1}$ 258 259 for the projectiles with densities ranging from 14.9 to 1.2 g cm⁻³. Both the diameter and 260 the depth of the craters became larger as the projectile density increased — except for the PC projectile with its different diameter. Fig. 3c has been re-scaled by using the $R_{\rm rim}$ 261 values in order to show the normalized crater profiles in Fig. 3d. The normalized crater 262 263 profiles were very similar around only the crater rim, regardless of the projectile density, 264 but the normalized crater depth was deeper as the projectile density increased. 265 We analyzed the normalized crater profiles to determine the following parameters (Fig. 1c): the crater rim diameter (D_{rim}) , the crater diameter measured on the pre-shot surface 266 $(D_{\rm c})$, the crater depth from the rim peak to the crater bottom $(d_{\rm rim})$, and the crater depth 267 268 from the pre-shot surface to the crater bottom (d_c) . All of these parameters are explained 269 in Fig. 1c. The calculated ratio of the crater rim diameter to the crater diameter, $D_{\rm rim}/D_{\rm c}$, 270 was calculated to be 1.26 ± 0.04 , regardless of the impact conditions. The ratio of the crater depth to the crater diameter, d_c/D_c , is shown in Fig. 4. As mentioned above, the 271

 $d_{\rm c}/D_{\rm c}$ increased with the increase of the projectile density, but it did not systematically 272 depend on the impact velocity for each projectile. This d_c/D_c increase with the 273 274 projectile density could have been caused by the penetration of the projectile into the 275 target, because a high-density projectile penetrates deeper due to its greater momentum. 276 We were able to determine the precise crater diameter, D_c , by using the crater profiles obtained by the laser profiler, but the crater profile was not measured in all shots. The 277 278 crater rim diameter, $D_{\rm rim}$, was measured in all shots instead of the $D_{\rm c}$. We thus used the following equation, $D_c = D_{rim}/1.26$, to construct the crater size scaling law. The 279 280 experimental conditions and the crater size results are summarized in Table 2. To study the relationship between π_4 and π_R in Eq. (4), we plotted the data for the 281 282 same projectile diameter at a constant impact velocity (i.e., a constant π_2) in Fig. 5a. This 283 figure includes the data for seven projectiles (the exception is the PC projectile) with 284 different densities at the impact velocity of 2 and 4 km s⁻¹. The $\pi_{\rm R}$ values increased 285 slightly with the increase of π_4 at each impact velocity. This relationship can be described as follows: 286

287
$$\pi_{\rm R} = 10^a \cdot \pi_4{}^b.$$
 (5)

The coefficients of *a* and *b* were $a = 1.24 \pm 0.01$ and $b = 0.047 \pm 0.019$ for $v_i = 2$ km s⁻¹, and $a = 1.34 \pm 0.01$ and $b = 0.042 \pm 0.017$ for $v_i = 4$ km s⁻¹. The coefficient *b* was independent of the impact velocity, and its calculated average was 0.044. Therefore, the Table 2

291 power law index of π_2 in Eq. (4) can be determined from the relationship between $\pi_R \cdot$ 292 π_4^{-b} and π_2 in Fig. 5b as follows:

293
$$\pi_{\rm R} \cdot \pi_4^{-b} = 10^{-0.265 \pm 0.014} \pi_2^{-0.176 \pm 0.002}.$$
 (6)

The power law indices of the coupling parameters, μ and ν , in Eq. (4) were derived from *b* and the power law index of π_2 , and their values were $\mu = 0.43$ and $\nu = 0.35$. The obtained μ was almost the same as that summarized by Housen and Holsapple [2011] for sand targets based on the studies of Stöffler et al. [1975] and Cintala et al. [1999]. Housen and Holsapple [2003] suggested that the typical value of ν could be assumed to be 1/3. Our present finding is quite similar to this value.

300

301 *3.2. Impact-induced seismic waves*

302 We classified the impact-induced seismic waves observed in this study into two 303 categories depending on the distance from the impact point: (1) a single pulse-like wave 304 (Fig. 6a, b), and (2) a damped oscillation wave (Fig. 6c, d). The single pulse-like wave 305 usually appeared near the crater rim. Fig. 6a, b provides an example of a single pulse-like 306 wave obtained at the impact velocity of 5.2 km s⁻¹ (run #140530-3). The acceleration 307 history in the range of time up to 100 ms is illustrated in Fig. 6a, and Fig. 6b shows the 308 history up to 15 ms. The time of '0 ms' corresponds to the time at which the projectile 309 impacted the target surface, which we determined from the high-speed digital camera

310 observation.

311 A complex wave with a small amplitude and a short wavelength was typically observed after 50 ms from the impact (Fig. 6a). Our high-speed digital camera 312 313 observations revealed that many ejected grains from the crater dropped on the target 314 surface at ~ 50 ms after the impact. This complex wave could therefore also be generated 315 by the impact of grains dropped on the accelerometer. We then limited the seismic waves 316 to those that occurred during the first 30 ms for the analysis of these waveforms. The 317 single pulse-like wave was approximated by a damped oscillation wave with a large 318 damping rate, and the wave can thus be described as one cycle of a damped sinusoidal 319 curve with a frequency of several hundred Hz. The amplitude of the acceleration 320 attenuated with the propagation distance.

321 These features were very similar to those reported for an unconsolidated loose sand 322 target [McGarr et al., 1969] and those observed with a target of glass beads [Yasui et al., 323 2015]. The damped oscillation wave usually appeared far from the crater rim. Fig. 6c, d provide an example of the damped oscillation obtained at the impact velocity of 4.1 km 324 325 s^{-1} (run #150309–1); Fig. 6c shows the acceleration history in the range of time up to 200 326 ms, and Fig. 6d shows that up to 20 ms. A complex wave with a small amplitude and a 327 short wavelength was also observed after 100 ms of the impact (Fig. 6d). This complex 328 wave could also be generated by the impact of grains that fell on the accelerometer. The 329 distance between the accelerometer and the crater rim shown in Fig. 6c, d was greater

330 than the x = 15 cm of the accelerometer (Fig. 6a, b). It thus appears that the complex

331 wave was generated by grains that fell at a later time. The seismic wave observed far from

the crater rim is also approximated by a damped oscillation wave with a smaller damping

rate as shown in Fig. 6c, d; the wave typically oscillated several times.

334 We analyzed these observed seismic waves to obtain three parameters characterizing 335 them as expressed in Fig. 6e: the traveling time (t_{max}) , the maximum acceleration (g_{max}) , 336 and the duration of the first upward acceleration (T_{half}) . We used these parameters to 337 determine the decay rate of the acceleration during the wave propagation. The Table 3 experimental conditions and results including the data of three parameters are 338 Table 4 summarized in Tables 3–5. Table 3 concerns the PC projectile at the impact velocity of 339 Table 5 340 \sim 200 m s⁻¹, Table 4 is about the PC projectile at impact velocities >1 km s⁻¹, and Table 5 341 is the results for 2-mm projectiles with the projectile density ranging from 14.9 to 2.5 g

342 cm^{-3} at impact velocities >1 km s⁻¹.

343

344 3.3. Propagation velocity of impact-induced seismic waves

The traveling time of the impact-induced seismic wave (t_{max}) is defined as the duration between the impact time and the time point at which the maximum acceleration of the first upward peak is observed. We used t_{max} to determine the propagation velocity of each impact-induced seismic wave on the target. Fig. 7a illustrates the relationship between the traveling time t_{max} and the propagation distance, x, for a PC projectile at

350 the x from 5 cm to 30 cm at different impact velocities. The average crater rim radius at

ach impact velocity is also shown on this figure as a horizontal line and in Table 6.

The t_{max} value increases linearly with the increase of the x outside the crater rim 352 at each impact velocity, and thus the propagation velocity (V_{prop}) of the impact-induced 353 seismic waves could be constant outside the crater rim. We therefore calculated the V_{prop} 354 from the slope of this linear relationship, and each dataset corresponding to each impact 355 356 velocity was fitted by the linear equation. The V_{prop} obtained at each impact velocity is 357 shown in Table 6. Regardless of the impact velocity in the range from 7 to 0.2 km s^{-1} , the V_{prop} values were almost constant within the error. The average V_{prop} was 51.1 ± 8.9 m 358 s⁻¹. The traveling time, t_{max} , should be zero at the impact point of x = 0, but we 359 observed that all of the fitting lines had offsets at the intersection on the vertical axis at 360 361 t = 0 (Fig. 7a). The offset was larger as the impact velocity was higher. This offset was 362 also reported by Yasui et al. [2015], who suggested that the offset might be explained by 363 a high propagation velocity of an impact-induced seismic wave inside the crater rim.

At high-velocity impacts, the target around the impacted region was highly compressed by a shock wave to achieve high pressure [Melosh, 1989], and the shock wave propagated inside the crater. A shock wave velocity is experimentally known to increase with the shock pressure, and therefore the traveling time inside the crater could be dominated by the shock wave velocity, and then the large offset might be explained by the short traveling time of the shock wave inside the crater. The t_{max} at the crater rim Table 6

370 for each impact velocity can be obtained by extrapolating each fitting line.

By using the t_{max} at the crater rim and the impact time ($t_{\text{max}} = 0$ at x = 0), we 371 372 can speculate that the propagation velocity of an impact-induced seismic wave inside the 373 crater rim is about three times higher than that outside the crater rim, ~ 175 m s⁻¹. This 374 high velocity could be caused in part by the shock wave propagation inside the crater rim. Fig. 7b delineates the relationship between the t_{max} and the x for 2-mm projectiles 375 376 with different densities at the impact velocity of 4 km s⁻¹. We suspected that this 377 relationship might move upward as the average crater rim radius becomes larger, and this 378 behavior was recognized at the other impact velocities. However, the slope of the linear fitting line was almost the same, regardless of the projectile density. The V_{prop} obtained 379 for each projectile at each impact velocity is listed in Table 7. The average V_{prop} was 380 381 53.0 ± 6.4 m s⁻¹, which is almost the same as that for the PC projectile at different impact 382 velocities.

Table 7

Yasui et al. [2015] measured the propagation velocity of impact-induced seismic waves in glass beads with the mean diameter of 200 μ m: 108.9 ± 16.2 m s⁻¹, which is twice the velocity for the quartz sand obtained in the present experiments. They discussed the effect of the frequency of the seismic wave on the propagation velocity, and they concluded that the propagation velocity decreased with the decrease of the frequency. In the present investigation, the measured frequency of the first peak was 700–100 Hz, which is approximately one-half of the values reported by Yasui et al. [2015]. Thus, this 390 might be one of the reasons that the propagation velocity of quartz sand is smaller than 391 that of glass beads. In addition, the grains of the quartz sand used herein have irregular 392 shapes, differing sizes, and various angles of repose; these differences might also 393 contribute to the low propagation velocity.

394

395 *3.4. Maximum acceleration of impact-induced seismic waves*

396 To examine the attenuation process of impact-induced seismic waves, we measured

397 the maximum acceleration of the impact-induced seismic wave (g_{max}) in our experiments.

Fig. 8a clarifies the relationship between the g_{max} and the x for a PC projectile at the impact velocities from 7 to 0.2 km s⁻¹; the decay rate of the acceleration with the distance should be derived from this relationship. It is quite clear in the figure that the g_{max} attenuates with the x at each impact velocity, and the measured g_{max} was from 100 to 1 m s⁻². Thus, the g_{max} values can be fitted by the following power law equation:

403
$$g_{\max} = 10^{a_{r}} \cdot x^{-b_{r}},$$
 (7)

404 where a_r and b_r are constants, and b_r represents the attenuation rate of an impact-405 induced seismic wave. These values are provided in Table. 6. The slope b_r at the impact 406 velocities from 7 to 4 km s⁻¹ shows similar values from 3.1 to 2.5, whereas the b_r at the 407 impact velocities from 3 to 0.2 km s⁻¹ changed with the impact velocity and/or the 408 distance from the impact point, from 3.8 to 1.1. In contrast, the a_r , which is related to the

409 magnitude of the absolute acceleration, depended to a slight extent on the impact velocity410 as shown in Table. 6.

411 Fig. 8b shows the relationship between the g_{max} and the x obtained for different projectiles at the constant impact velocity of 4 km s⁻¹. The measured g_{max} ranged from 412 413 40 to 1 m s⁻². Each data set for the different projectiles was fitted by Eq. (7). The obtained $a_{\rm r}$ and $b_{\rm r}$ are shown in Table 7. The $a_{\rm r}$ value increased with the increase of the 414 415 projectile density, whereas the b_r value systematically changed with the projectile 416 density and was between 1.9 and 3.1. The obtained a_r and b_r values at different impact 417 velocities showed a similar relationship with the projectile density (Table 7). For each 418 projectile, the obtained a_r increased with the increase of the impact velocity, similar to 419 that observed for the PC projectile.

420 Yasui et al. [2015] clarified that the g_{max} was scaled well by using the distance from 421 the impact point normalized by the crater rim radius (x/R_{rim}) because the R_{rim} includes 422 all of the effects of the impact conditions according to the crater size scaling law described 423 in Eq. (6). This relationship between g_{max} and x/R_{rim} can be written as follows:

424
$$g_{\rm max} = 10^{a_{\rm n}} (x/R_{\rm rim})^{-b_{\rm n}}.$$
 (8)

Fig. 8c-e provides the g_{max} results expressed by the normalized distance, x/R_{rim} . Panels (c) and (d) of Fig. 8 illustrate the data for a PC projectile and an Al projectile at different impact velocities, respectively, and Fig. 8e is for all of the projectiles with 428 different projectile densities at different impact velocities. As seen in Fig. 8c, the data for 429 the 4.75-mm-dia. PC projectile are slightly scattered, and the data at 1 km s⁻¹ are 430 somewhat smaller values than the others, but all of the g_{max} values obtained at different 431 impact velocities were well merged and could be scaled by the normalized distance.

432 Similarly, Fig. 8d (for a 2-mm-dia. Al projectile) demonstrates that all of the data were merged well, and the relationship between g_{max} and the normalized distance was 433 434 independent of the impact velocity. The data for the other 2-mm-dia. projectiles showed 435 a similar tendency. Fig. 8e indicates that all of the data for the different projectiles merged 436 well, and this relationship was independent of both the impact velocity and the projectile density. The data could be fitted by using Eq. (8), and the a_n and b_n were 2.21 ± 0.04 437 438 and 3.18 ± 0.10 , respectively. From the empirical equation, the g_{max} at the crater rim 439 $(x/R_{\rm rim} = 1)$ was estimated as 160 m s⁻²; this value is approximately 16 times larger than the Earth's gravity. At $g_{\text{max}} = 1$ G, the normalized distance was estimated as 2.4. 440 441 The quartz sand in this region might have been fluidized by the impact-induced seismic 442 wave.

Yasui et al. [2015] measured the g_{max} of 200-µm glass beads for PC, alumina, and SUS projectiles at impact velocities <150 m s⁻¹. The empirical equation fitted by Eq. (8) is also shown in Fig. 8e. Comparing our results with those reported by Yasui et al., the decay rate (b_n) of the quartz sand was approximately 1.5 times larger than that of the glass beads. Surprisingly, the g_{max} of the quartz sand at the crater rim ($x/R_{\text{rim}} = 1$) 448 was almost the same as that of the glass beads, i.e., $\sim 150 \text{ m s}^{-2}$. We thus speculated that 449 the minimum acceleration necessary for crater excavation at the crater rim might be 450 approximately 15 G in the gravity regime for a crater formation process at the crater 451 diameters from 20 to 6 cm in this experiment. An excess acceleration beyond 1 G 452 corresponding to the fluidization of granular materials might be necessary to cause 453 excavation flow inside the crater rim.

454

455 *3.5. Duration of first upward acceleration*

The duration of the first upward acceleration could be explained by the generation mechanism of impact-induced seismic waves. Yasui et al. [2015] defined the duration of the first upward acceleration as a half-period of the first wave (T_{half}), which indicates the length of time between the time point at which the acceleration rose from the background level and the time point at which it fell back down to the background level.

461 Fig. 9a, b illustrates the relationship between the T_{half} and the distance from the 462 impact point: Fig. 9a is for a PC projectile at different impact velocities, and Fig. 9b is 463 for 2-mm-dia. projectiles with different densities at the impact velocity of 4 km s⁻¹. In 464 Fig. 9a, the data of the T_{half} are somewhat scattered but they are independent of the 465 distance from the impact point. The T_{half} results were somewhat dependent on the 466 impact velocity from 7 to 1 km s⁻¹, and the T_{half} at 200 m s⁻¹ was clearly smaller than 467 the values at high impact velocities. Yasui et al. [2015] measured the T_{half} of 200-μm

468 glass beads at impact velocities $<150 \text{ m s}^{-1}$. They reported a T_{half} of $\sim 0.7 \text{ ms}$, which is 469 slightly smaller than that at 200 m s⁻¹ and much smaller than the values obtained at high 470 impact velocities in our present experiments.

As shown in Fig. 9b, the T_{half} was independent of both the distance from the impact 471 point and the projectile density. The average T_{half} for the 2-mm-dia. projectiles at 4 km 472 s^{-1} was 2.01 ± 0.38 ms, which is approximately three times greater than that of glass beads. 473 474 In light of the data in these figures, we can say that the T_{half} increases with the increase 475 of the impact velocity, and it is independent of the projectile density. Our examination of 476 this tendency revealed the relationship between the T_{half} and the impact velocity for 477 each projectile (Fig. 9c). The data are slightly scattered and the error was not very small 478 for each result, but the T_{half} increased exponentially with the increase of the impact 479 velocity. Moreover, all of the data could be fitted by one empirical power law equation as 480 follows:

481
$$T_{\text{half}} [\text{ms}] = (1.65 \pm 0.11) \cdot (v_i [\text{km s}^{-1}])^{0.14 \pm 0.05}.$$
 (9)

482 Yasui et al. [2015] suggested that the T_{half} could be consistent with the duration of a 483 projectile penetration into the target. They calculated the penetration duration of the 484 projectile (t_p) by using the deceleration model proposed by Niimi et al. [2011]. They 485 chose the resistance law described by the dynamic pressure for a deceleration mechanism 486 in a hydrodynamic regime: $t_p \propto v_i^{-1}$. However, our experimental results cannot be 487 explained by this resistance law, because T_{half} is proportional to $v_i^{0.14}$. We thus chose 488 another resistance law described by viscous drag force, as follows:

$$m_{\rm p}\frac{dv}{dt} = -A_0\eta r_{\rm p}v,\qquad(10)$$

490 where v is the projectile penetration velocity, t is the time, η is the viscosity of target 491 material, and A_0 is the constant related to the geometrical factor determined by an 492 experiment. The penetration duration of the projectile (t_p) could be obtained by the 493 following equations from the solution of Eq. (10) at $v = v_0$ (constant):

494
$$t_{\rm p} = -\tau \ln\left(\frac{v_0}{v_{\rm i}}\right) \tag{11}$$

495
$$\tau = \frac{4\pi \rho_{\rm p} r_{\rm p}^{\ 2}}{3A_0 \eta}.$$
 (12)

The fitting line obtained by using Eq. (11) is also shown in Fig. 9c, and it can be seen that Eq. (11) agrees with all of the data, as does Eq. (9). We can obtain the penetration velocity just before the projectile stops (v_0) and the effective viscosity ($A_0\eta$) from Eqs. (11) and (12): the v_0 was 1.8 m s⁻¹, regardless of the projectile properties, and the $A_0\eta$ increased with the increase of the projectile density for the same projectile radius and the value was 40–240 Pas.

502

503 **4. Discussions**

504 *4.1. Impact-induced seismic energy propagating with distance*

505 The impact-induced seismic energy (E_s) is an important factor in evaluations of the

seismic shaking induced by the impact of small bodies on solid planets and asteroids. In this section, we estimate the E_s by using the empirical equations we obtained herein that describes the three parameters V_{prop} , g_{max} , and T_{half} . In the model proposed by Yasui et al. [2015], it is assumed that the E_s is the kinetic energy of the target material vibrating at the shell region, with a width corresponding to one cycle of the impact-induced seismic wave. The impact-induced seismic energy E_s is determined as follows:

512
$$E_{\rm s} = \int_{x_1}^{x_2} \frac{1}{2} \cdot 2\pi x^2 \rho_{\rm t} V(x,t)^2 dx, \qquad (13)$$

where ρ_t is the target density, x_1 and x_2 are the inside and outside distances of the shell region from the impact point, t is the time, and V(x, t) is the vibration velocity of the target particle induced by the impact. For the estimation of E_s , it is difficult to use the original waveform of the acceleration obtained in the present study because the waveform had a complex shape, as shown in Fig. 6. Therefore, in order to calculate the V(x, t), we simulated the impact-induced seismic wave (Fig. 10a).

The seismic wave has a positive phase and a negative phase (see Fig. 6), and a positive acceleration phase has a larger amplitude compared to a negative acceleration phase. We therefore divided the seismic wave into two regions: one region showing a half cycle of a sinusoidal wave with T_{half} and g_{max} in the positive amplitude region, and the other region with T_{2half} and g_{min} in the negative amplitude region. The acceleration in each region is expressed as follows:

525
$$a_{\rm up}(x,t) = g_{\rm max} \sin\left[\frac{\pi}{T_{\rm half}} \left(t - \frac{x}{V_{\rm prop}}\right)\right], \qquad (14)$$

526
$$a_{\text{down}}(x,t) = g_{\min} \sin\left[\frac{\pi}{T_{\text{2half}}}\left\{t - \left(T_{\text{half}} + \frac{x}{V_{\text{prop}}}\right)\right\}\right], \quad (15)$$

where $a_{up}(x,t)$ and $a_{down}(x,t)$ are the acceleration in the positive and negative 527 amplitude regions, respectively. The average V_{prop} obtained was 52.4 ± 7.2 m s⁻¹, 528 regardless of the impact velocity and the projectile property. The g_{max} depended on the 529 distance from the impact point normalized by the crater rim radius, x/R_{rim} , and it is 530 expressed as Eq. (8). The g_{\min} showing the maximum acceleration in the negative 531 532 amplitude region could be obtained from our experimental results, and it is illustrated in 533 Fig. 10b; the relationship between the absolute value of $g_{\rm max}/g_{\rm min}$ and the $x/R_{\rm rim}$ is 534 shown. The $|g_{\text{max}}/g_{\text{min}}|$ had a good correlation with the x/R_{rim} , regardless of the 535 projectile density and the impact velocity, and it can be expressed as the following power 536 law equation:

537
$$|g_{\text{max}}/g_{\text{min}}| = 10^{1.09 \pm 0.03} (x/R_{\text{rim}})^{-2.20 \pm 0.07}.$$
 (16)

538 The T_{half} was slightly dependent on the impact velocity (v_i) ; it is expressed as Eq. (9). 539 The vibration velocity of the target particle can be obtained by calculating the integral of 540 Eqs. (14) and (15) for t, and it can be expressed as follows:

541
$$V_{\rm up}(x,t) = \frac{T_{\rm half}}{\pi} g_{\rm max} \left[1 - \cos\left\{ \frac{\pi}{T_{\rm half}} \left(t - \frac{x}{V_{\rm prop}} \right) \right\} \right], \tag{17}$$

542
$$V_{\text{down}}(x,t) = \frac{2T_{\text{half}}}{\pi}g_{\text{max}}$$

543
$$+ \frac{T_{2\text{half}}}{\pi} g_{\min} \left[1 - \cos\left\{ \frac{\pi}{T_{2\text{half}}} \left(t - \left\{ T_{\text{half}} + \frac{x}{V_{\text{prop}}} \right\} \right) \right\} \right].$$
(18)

Here, the T_{2half} was determined by the following boundary condition: $V_{down}(x, x/V_{prop} + T_{half} + T_{2half}) = 0$. From this boundary condition, the T_{2half} can be expressed as:

547
$$T_{\text{2half}} = -T_{\text{half}} \left(\frac{g_{\text{max}}}{g_{\text{min}}}\right), \quad (19)$$

548 and Eq. (18) can be rewritten as follows:

549
$$V_{\text{down}}(x,t) = \frac{T_{\text{half}}}{\pi} g_{\text{max}} \left[1 + \cos\left\{ -\frac{\pi}{T_{\text{half}}} \left(\frac{g_{\text{max}}}{g_{\text{min}}} \right) \left(t - \left\{ T_{\text{half}} + \frac{x}{V_{\text{prop}}} \right\} \right) \right\} \right].$$
(20)

Fig. 11a-c presents an example of the original waveform obtained in this experiment 550 551 and the simulated waveforms calculated by using the Eqs. (14), (15), (17), and (20) for the 2-mm-dia. Al projectile at the impact velocity of 4.14 km s⁻¹ (run #151105–3). Fig. 552 553 11d clarifies the relationship between the vibration velocity, V(x, t), and the distance from the impact point normalized by the crater rim radius x/R_{rim} at different times from 554 18.0 to 7.5 ms. The maximum V(x,t) decreased with the increase of the time. In 555 556 addition, the normalized distance at the maximum V(x,t), i.e., $x_{\text{peak}}/R_{\text{rim}}$, increased 557 with the increase of the time.

558 The x_1 and x_2 values of Eq. (13) can be obtained from this relationship: they can 559 be calculated at the V(x,t) = 0 for each time. The impact-induced seismic energy (E_s) 560 can be calculated by substituting Eqs. (17) and (20) into Eq. (13) with x_1 and x_2 . Fig.

12a shows the relationship between the impact-induced seismic energy normalized by the projectile kinetic energy, E_s/E_k , and the normalized distance at the maximum V(x, t), i.e., $x_{\text{peak}}/R_{\text{rim}}$, in the range of $x_{\text{peak}}/R_{\text{rim}}$ from 1 to 10 at various impact conditions. The E_s/E_k depends on the impact velocity and the projectile size and density. For example, the E_s/E_k for a PC projectile at 200 m s⁻¹ was more than 10 times larger than that at 7 km s⁻¹, and the E_s/E_k for an Al projectile (2.7 g cm⁻³) was twice as large as that for a WC projectile (14.9 g cm⁻³) at 4 km s⁻¹.

The E_s/E_k continued to decrease with the increase of the normalized distance 568 $x_{\text{peak}}/R_{\text{rim}}$, following the power law relationship of $E_s/E_k \propto (x_{\text{peak}}/R_{\text{rim}})^{-4.5}$. This 569 continuous decrease of the E_s/E_k could be caused by the energy dissipation in the plastic 570 571 wave. The impact generates a detached stress wave, initially as a strong shock wave, that 572 propagates away from the impact point, decays in magnitude, and eventually becomes an 573 elastic seismic wave. When the stress wave is in the shock and plastic regimes it loses the 574 energy rapidly with the normalized distance. On the other hand, once the stress wave 575 becomes an elastic wave, the energy losses should be small.

576 Thus, the elastic seismic wave did not appear in these experimental conditions, and 577 the impact-induced seismic efficiency factor could not be determined because the impact-578 induced seismic efficiency factor is the E_s/E_k that ends up in the seismic/elastic wave 579 that can propagate large distances from the impact point. We expect that the E_s/E_k 580 continues to decrease until the plastic wave changes to the elastic wave, and then the 581 E_s/E_k becomes constant. Therefore, our obtained E_s/E_k could be the upper limit of the 582 impact-induced seismic efficiency factor.

583 We compared the E_s/E_k values obtained in this study with the reported impact-584 induced seismic efficiency factor from previous experiments. For example, McGarr et al. [1969] examined the seismic impulse normalized by the kinetic energy of a projectile, 585 $I/E_{\rm k}$ (which is similar to our $E_{\rm s}/E_{\rm k}$ parameter) for a loose sand target at impact 586 velocities from 7 to 2 km s⁻¹ in laboratory experiments. They obtained the average $I/E_{\rm k}$ 587 value of $(0.6 \pm 0.4) \times 10^{-5}$. We replotted their I/E_k values against the normalized 588 589 distance shown in Fig. 12a: we calculated the crater rim radius for each experiment in the 590 paper of McGarr et al. [1969] by using our crater size scaling law expressed by Eq. (6). 591 The McGarr et al. data then seemed to increase with the increase of the normalized 592 distance; this trend is the opposite of our results for each impact condition. As we described above, the E_s/E_k depends on the impact velocity, and it could increase with 593 594 the decrease of the impact velocity.

We thus presume that the increase of the I/E_k with the increase of the normalized distance is not a true dependency on the distance; rather, it could be caused by the impact velocity change from 7 to 2 km s⁻¹. If so, it might be possible that the I/E_k also decreases with the increase of the normalized distance, because the maximum acceleration g_{max} was observed to decrease with the normalized distance, x/R_{rim} , for the loose sand used by McGarr et al. [1969] as shown in Fig. 12b. That is, their obtained seismic wave did not also appear to be the elastic wave. Their obtained I/E_k might therefore correspond to the upper limit of the impact-induced seismic efficiency factor, as is the case in our present investigation. Their I/E_k value was larger than the values we obtained herein, even though the impact velocity range was almost the same. The g_{max} for loose sand is also greater than that for our quartz sand (Fig. 12b). This might be caused by differences in mechanical properties of the target, such as the bulk sound velocity and the target density.

608 Moore et al. [1970] carried out large-scale impact experiments using a missile at the estimated impact velocity of 2.5 km s⁻¹ at the White Sands Missile Range (New Mexico, 609 U.S.). The impact-induced seismic efficiency factor of 10^{-5} at the quite large kinetic 610 energy of the projectile, 4.5×10^{20} J, was obtained. In addition, Latham et al. [1970] 611 estimated the impact-induced seismic efficiency factor as 10^{-5} to 10^{-6} by using the lunar 612 613 module impact provided by the Apollo; the impact velocity of the lunar module was estimated to be 1.7 km s⁻¹ and the kinetic energy was 3.6×10^{11} J [Toksöz et al., 1974]. 614 615 Their values are larger than our calculated E_s/E_k at the normalized distances far from 6; 616 however, our values could be the upper limit of the impact-induced seismic efficiency 617 factor.

618 Herein, we measured the acceleration along the direction perpendicular to the target 619 surface. However, the particle motion induced by the seismic wave has not only a vertical 620 component (Z); the motion also has horizontal (X) and radial (Y) components in the target

Fig. 13

621 (Fig. 13a). We thus measured the acceleration excited along these three axes by using a 622 triaxial accelerometer to confirm the amplitude of the acceleration along the X- and Y-623 axes. Fig. 13b shows the acceleration measured at the X-, Y-, and Z-axes for a PC 624 projectile at 2.4 km s⁻¹. The triaxial accelerometer was set at 11 cm from the impact point 625 $(x/R_{rim} = 1.5)$. We observed that acceleration at the Y-axis was not detected.

626 Based on the crater formation model proposed by Maxwell [1977], each flow line 627 drawn from an impact point never crosses another flow line, which means that the particle 628 motion does not have a radial component. This is the reason why the acceleration at the 629 Y-axis was not detected. In addition, we firstly observed that the maximum acceleration at the X-axis was about five times larger than that at the Z-axis, and thus it is clear that 630 the kinetic energy of the particle is dominated by the horizontal motion. When we 631 632 calculate the particle velocity considering the acceleration along the X-axis, the $E_{\rm s}/E_{\rm k}$ could be more than one order of magnitude larger than that calculated in Fig. 12a. 633 634 This is why our calculated E_s/E_k was smaller than the impact-induced seismic efficiency factor estimated from the missile experiments and the lunar module impact 635 636 [Moore et al., 1970; Latham et al., 1970]. In the future, in order to determine the impact-637 induced seismic efficiency factor, it is necessary to measure the acceleration by using a triaxial accelerometer at a distant region where the seismic wave changes to the elastic 638 639 wave.

640

641 4.2. Possibility of detecting an impact-induced seismic shaking on asteroids

642 To estimate the effects of impact-induced seismic shaking on asteroids, we used the 643 empirical equations that we obtained in this study. These equations were obtained under 644 Earth's gravity, 1 G, so we should consider the effect of microgravity when applying our 645 established model to the resurfacing processes that are due to impact-induced seismic 646 shaking on asteroid surfaces. As described next, we constructed an impact-induced 647 seismic shaking model including the effect of gravity on asteroids by comparing our 648 experimental results with the scaling law for ejecta velocity distribution, and we discuss 649 the possibility of the impact-induced seismic shaking on asteroids.

We first consider the particle velocity ejected at the position of the crater radius in accord with the scaling law for ejecta velocity distribution that was proposed by Housen and Holsapple [2011]. The ejection velocity at the position of the crater radius, v_e , is written as follows:

654
$$v_{\rm e} = c_0 \sqrt{g R_{\rm c}},$$
 (21)

where *g* is the gravitational acceleration in m s⁻² and c_0 is the constant determined by experiments for each target material. Next, we assume that the ejection velocity of a particle at the position of the crater radius ($x/R_c = 1$) can be simply estimated by using the product of the maximum acceleration, g_{max} , in Eq. (8), and the duration of the first upward acceleration, T_{half} . The maximum acceleration at $x/R_c = 1$ is rewritten as
660 $g_{\text{max}} = (1.26)^{b_n} g_0$ under a certain gravity, where $R_c = R_{\text{rim}}/1.26$ and g_0 is the 661 maximum acceleration at the crater rim. Thus, the ejection velocity can be written as 662 follows:

663
$$v_{e\perp}[m \, s^{-1}] = g_{max} \cdot T_{half} = (1.26)^{b_n} g_0 T_{half},$$
 (22)

where $v_{e\perp}$ is the ejection velocity of the particles in the normal direction. It is assumed that the target particles are ejected from the target surface with the ejection angle of 45°, a typical ejection angle of granular materials [e.g., Tsujido et al., 2015], and it is assumed that the ejection velocity estimated from the scaling law of Eq. (21) is equal to be that obtained in this experiment as Eq. (22). Thus, the ejection velocity of the particles at $x/R_c = 1$ can be rewritten as $v_e = \sqrt{2}v_{e\perp}$, and the following equation is constructed by using Eqs. (21) and (22):

671
$$c_0 \sqrt{gR_c} = \sqrt{2} \times (1.26)^{b_n} g_0 T_{\text{half}},$$
 (23)

672 that is,

673
$$g_0 = \frac{c_0 (1.26)^{-b_n}}{\sqrt{2}} \times \frac{\sqrt{gR_c}}{T_{\text{half}}}.$$
 (24)

674 The R_c is calculated from the crater size scaling law of Eq. (6) and is shown as follows:

675
$$R_{\rm c} = (8.75 \times 10^{-1})g^{-0.176} v_{\rm i}^{0.352} r_{\rm p}^{0.824} \left(\frac{\rho_{\rm t}}{\rho_{\rm p}}\right)^{-0.289}, \quad (25)$$

676 where the projectile mass, m_p , can be written as $4\pi\rho_p r_p^3/3$ by assuming a spherical

677 shape. Therefore, the g_0 can be rewritten as follows:

678
$$g_0 = \frac{(3.17 \times 10^{-1})c_0 (gr_p)^{0.412} v_i^{0.176} (\rho_t / \rho_p)^{-0.145}}{T_{\text{half}}}, \quad (26)$$

where $b_n = 3.18$ in Eq. (8). To compare our semi-theoretical equation for g_0 with the obtained g_0 , which is 10^{a_n} ($a_n = 2.21 \pm 0.04$) in Eq. (8), in our experiments, we used the T_{half} of the empirical equation, Eq. (9). Substituting Eq. (9) into Eq. (26), the g_0 can be rewritten as follows:

683
$$g_0 = (3.97 \times 10^2) (gr_p)^{0.412} v_i^{0.036} \left(\frac{\rho_t}{\rho_p}\right)^{-0.145}, \qquad (27)$$

where $c_0 = 0.76$ m s⁻¹ as obtained by Tsujido et al. [2015]. Fig. 14 shows the 684 calculated results of Eq. (27) at the impact velocity (v_i) of 0.2–7 km s⁻¹ for various 685 686 projectiles. The gray zone shown in Fig. 14 represents the 10^{a_n} in Eq. (8), and all of the calculated g_0 values were a little smaller than the observed g_0 (10^{*a*_n} in Eq. (8)). The 687 difference between the calculated g_0 values and the observed g_0 might be caused by 688 689 the ejection angle and/or the duration, T_{half} . The g_0 was measured below the target surface (2.5 cm from the target surface) in our experiments while the calculated g_0 was 690 estimated on the target surface. Therefore, the g_0 might depend on the depth from the 691 692 target surface. However, their differences were within a factor of 2 and they were 693 consistent within error bars. Our semi-theoretical equation of Eq. (26) could be suitable 694 for the estimation of g_0 at various impact conditions.

Fig. 14

Although the g_{max} at the crater rim, that is, the g_0 was very similar in all of our experiments, the semi-theoretical equation of Eq. (26) suggests that the g_0 is sensitive to the product of surface gravity and the projectile radius. In our experiments, the projectile radius was changed by a factor of 2, and thus an effect of the projectile radius on the g_0 was not recognized. This result should be verified by future experiments and/or numerical simulations.

Lastly, we considered the impact-induced seismic shaking on an idealized small (500 m in diameter) body with a bulk density of 1.46 g cm⁻³, which is the same as our target. In this calculation, we should include the effect of the impactor size on the T_{half} of Eq. (26), and we therefore used the theoretical equations of Eqs. (11) and (12). Substituting Eqs. (11) and (12) into Eq. (26), the g_0 can be rewritten as follows:

706
$$g_0 = -(5.79 \times 10^{-2})(A_0\eta)(gr_p)^{0.412} v_i^{0.176} r_p^{-2} \rho_t^{-0.145} \rho_p^{-0.856} \left[\ln\left(\frac{v_0}{v_i}\right) \right]^{-1}$$
, (28)

where $c_0 = 0.76$ m s⁻¹ obtained by Tsujido et al. [2015], and $A_0\eta$ and v_0 were determined by our experiments using PC projectiles. The $A_0\eta$ and v_0 values were 114 Pas and 1.3 m s⁻¹, respectively. In this calculation, the impactors with the diameter of 0.01, 0.1, 0.5, and 1 m and the density of 1.4 g cm⁻³ (which is the average density of Ctype asteroids as shown by Britt et al. [2002]) were assumed to collide at the impact velocity of 5 km s⁻¹, a typical collisional velocity in the region of main belt. We then calculated the maximum acceleration attenuated with the distance calculated for the 714 different sizes of the impactor by using Eq. (8), i.e., $g_{\text{max}} = g_0 (x/R_{\text{rim}})^{-3.18}$.

715 Fig. 15 illustrates the relationship between the distance from the impact point normalized by the crater rim radius, x/R_{rim} , and the maximum acceleration, g_{max} . The 716 717 region at which the impact-induced seismic shaking occurred was assumed to be within 718 the distance from the impact point corresponding to the g_{max} beyond the surface gravity 719 of the idealized small body. The g_{max} was observed to increase with the decrease of the 720 impactor size at the same normalized distance. The g_{max} with the impactor size of 0.01 m at $x/R_{\rm rim} = 1$ was more than three orders of magnitude larger than that obtained with 721 722 a 1-m impactor.

723 The area at which the impact-induced seismic shaking occurred changed with the 724 impactor size. The normalized distance at which the impact-induced seismic shaking 725 occurred changed from 9 to 0.9 when the impactor size increased from 0.01 m to 1 m. 726 When an impact crater with a rim diameter of 2.8 m is formed by a 0.01-m impactor on 727 this idealized small body, impact-induced seismic shaking could be induced within the 728 distance of 12 m from the impact point. However, when a crater with a diameter of 123 729 m is formed by a 1-m impactor, the region at which the impact-induced seismic shaking 730 occurs could be limited to within the crater rim, and thus the seismic shaking will not 731 occur far from the crater.

732

5. Summary

Fig. 15

734		We carried out impact cratering experiments using eight types of spherical projectiles
735	wit	h density ranging from 14.9 to 1.2 g cm ^{-3} on 500- μ m quartz sand at the impact velocity
736	fro	m 7 to 0.2 km s ^{-1} . We examined the waveform of the acceleration generated just after
737	the	impact, and we assessed the characteristics of the impact-induced seismic wave by
738	usi	ng three to five accelerometers set at 5.1 to 30.1 cm from the impact point in order to
739	con	sider the possibility of impact-induced seismic shaking on asteroids. Our results are
740	sun	nmarized as follows:
741	1.	We applied our results obtained under various impact conditions (e.g., the projectile
742		size, projectile density, and impact velocity) to the crater size scaling law proposed
743		by Housen and Holsapple [2011], and we determined the following empirical
744		equation, $\pi_{\rm R} \cdot \pi_4^{-0.044} = 10^{-0.265} \pi_2^{-0.176}$, where $\pi_{\rm R}$, π_2 , and π_4 are the non-
745		dimensional scaling parameters related to the crater rim radius, gravity, impact
746		velocity, projectile density, and target density. We obtained the power law indices of
747		the coupling parameters, μ and ν , as 0.43 and 0.35, respectively.
748	2.	The propagation velocity of an impact-induced seismic wave, V_{prop} , was determined
749		by using the duration between the impact time and the time showing the maximum
750		acceleration of the first upward peak. The obtained $V_{\rm prop}$ was 52.4 ± 7.2 m s ⁻¹
751		outside the crater, regardless of the impact velocity and the projectile properties. The
752		$V_{\rm prop}$ inside the crater rim was speculated to be ~175 m s ⁻¹ , which is approximately
753		three times larger than that outside the crater rim.

3. The maximum acceleration of the impact-induced seismic wave, g_{max} , could be scaled as the distance from the impact point normalized by the crater rim radius, x/R_{rim} , regardless of the impact velocity and the projectile properties, and the obtained empirical equation was shown as $g_{\text{max}} = 10^{2.21} \cdot (x/R_{\text{rim}})^{-3.18}$. The g_{max} of quartz sand at the $x/R_{\text{rim}} = 1$ obtained in this study matched well with that of the 200-µm glass beads reported by Yasui et al. [2015].

4. The duration of the first upward acceleration was defined as a half period of the first upward wave, T_{half} , and we observed that T_{half} was slightly dependent on the impact velocity but independent of the projectile properties. The obtained empirical equation was T_{half} [ms] = 1.65(v_i [km s⁻¹])^{0.14}. Our obtained T_{half} could be explained by the resistance law described by viscous drag force.

5. We calculated the seismic energy normalized by the kinetic energy of the projectile, E_s/E_k , by using our obtained empirical equations at the normalized distance from $x/R_{rim} = 1$ to 10 under various impact conditions. The E_s/E_k depended on the impact velocity and projectile properties, but the decay constant was approximately 4.5, regardless of the impact condition.

6. We estimated the distance from the crater rim at which the impact-induced seismic shaking occurred on an idealized small (500 m in diameter) body by using our semitheoretical equation. When we assumed the impact velocity of the impactor to be 5 $km s^{-1}$, the impact-induced seismic shaking area could be calculated for various impactor sizes.

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	11	0	1	1	1 2		5	

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783

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853	Figure	Captions:
000		captions

855	Figure 1
856	Schematic illustration of the experimental setup. (a) Two-stage vertical gas gun. (b) Five
857	accelerometers set on the target surface. (c) Cross-section of the accelerometers set on the
858	target surface, and an illustration of the crater dimensions.
859	
860	Figure 2
861	Impact craters on the quartz sand target. (a) The PC projectile at 1.13 km s ⁻¹ , run
862	#151104–3. (b) The PC projectile at 6.89 km s ^{-1} , run #141003–2, which was the largest
863	impact crater in this study. (c) The Al projectile at 4.12 km s ^{-1} , run #150309–1. (d) The
864	Cu projectile at 3.63 km s ^{-1} , run #150820–5. Each scale bar indicates 5.0 cm. The thick
865	black lines around the crater rim are the accelerometers' cords.
866	
867	Figure 3
868	Crater profiles obtained at different impact conditions. (a) The results obtained with the
869	PC projectile at the impact velocities of 1.1, 2.2, 4.2, and 6.2 km s ⁻¹ . (b) The crater
870	profiles expressed by the relationship between the values of x/R_{rim} and the y/R_{rim} for
871	panel (a). (c) The results of the projectiles with different densities at the impact velocity

872 of ~ 4.0 km s⁻¹. (d) The crater profile expressed by the relationship between the $x/R_{\rm rim}$

873 and the $y/R_{\rm rim}$ for panel (c).

874

875	Figu	ire 4
876	The	relationship between the ratio of the crater depth to the crater diameter, d_c/D_c , and
877	the j	projectile density. Each symbol indicates an impact velocity range.
878		
879	Figu	ire 5
880	(a)	The relationship between the non-dimensional scaling parameter π_4 and the π_R
881		defined in Eqs. (1) and (2). The results were obtained by the experiments using seven
882		projectiles with different densities at a constant impact velocity of 2.0 km $\rm s^{-1}$
883		(circles) and 4.0 km s ^{-1} (squares). The solid and dashed lines represent the fitting
884		lines determined by using the power law equation of Eq. (5).
885	(b)	The relationship between the non-dimensional scaling parameter π_2 and the π_R .
886		π_4^{-b} (in this study, we obtained $b = 0.044$). The solid line represents the fitting line
887		determined by using the power law equation of Eq. (6).
888		
889	Figu	ire 6
890	(a)	A typical sample of the impact-induced seismic wave showing a single pulse-like
891		wave observed in this study. A PC projectile was impacted at the velocity of 5.16 km

 s^{-1} (run #140530–3). The three different curves represent the data recorded by three

893		accelerometers set at different distances from the impact point, x . Time 0 is the
894		impact time of the projectile.
895	(b)	Enlargement of the horizontal axis of panel (a) ranging from -2 to 15 ms.
896	(c)	An example of the impact-induced seismic wave showing a damped vibration wave
897		observed far from the crater rim. An Al projectile was impacted at the velocity of
898		4.12 km s ⁻¹ (run #150309–1).
899	(d)	Enlargement of the horizontal axis of panel (c) ranging from -2 to 20 ms.
900	(e)	The physical properties of an impact-induced seismic wave: the traveling time (t_{max}) ,
901		the maximum acceleration (g_{\max}) , and the duration of the first upward wave of the
902		acceleration (T_{half}) .
903		
904	Fig	ure 7
905	The	relationship between the traveling time (t_{max}) and the distance from the impact point,
906	<i>x</i> (a	a) for PC projectile at different impact velocities, and (b) for 2-mm projectiles with
907	diff	erent densities at the impact velocity of 4 km s ⁻¹ . Horizontal lines represent the
908	aver	rage crater rim radius at each impact velocity. The fitting lines were obtained by using

Figure 8

a linear function.

912 (a, b) The relationship between the maximum acceleration (g_{max}) and the distance from

913 the impact point, x (a) for PC projectile at impact velocities ranging from 7 to 0.2 914 km s⁻¹ and (b) for 2-mm projectiles with the projectile density ranging from 14.9 to 915 2.5 g cm⁻³ at the impact velocity of 4 km s⁻¹. Each line represents the fitting line 916 obtained by using Eq. (7).

917 (c–e) The relationship between the maximum acceleration (g_{max}) and the distance from 918 the impact point normalized by the crater rim radius, x/R_{rim} (c) for PC projectile 919 at impact velocities from 7 to 0.2 km s⁻¹, (d) for Al projectile at impact velocities 920 from 5 to 2 km s⁻¹, and (e) for all data. The solid and dashed lines in panel (e) 921 represent the fitting lines obtained by using Eq. (8) and the empirical equation used 922 for 200-µm glass beads obtained by Yasui et al. [2015], respectively.

923

924 Figure 9

923	(a, b) The relationship between the duration of the first upward acceleration (T_{half}) and
926	the distance from the impact point, x (a) for PC projectile at impact velocities from
927	7 to 0.2 km s ^{-1} and (b) for 2-mm projectiles with the projectile density from 14.9 to
928	2.5 g cm ^{-3} at the impact velocity of 4.0 km s ^{-1} . The solid line in panel (b), and the
929	dotted line and gray area in panels (a) and (b) represent the average values for all of
930	the data obtained in this study, that for the 200-µm glass beads used by Yasui et al.
931	[2015], and the standard deviation for each data set, respectively.

932 (c) The average T_{half} at each impact velocity for all projectiles. The solid and dashed

lines represent the fitting lines obtained by using Eqs. (9) and (11), respectively.

935	Figure 10
936	(a) Schematic of a simulated impact-induced seismic wave for Eqs. (14) and (15).
937	(b) The relationship between the absolute ratio of the maximum acceleration in the
938	positive amplitude region to that in the negative amplitude region, i.e., $ g_{\text{max}}/g_{\text{min}} $,
939	and the normalized distance, x/R_{rim} for various projectiles. The solid line represents
940	the fitting line obtained with Eq. (16).
941	
942	Figure 11
943	(a) Waveforms of acceleration measured at $x = 10.7, 13.0, 15.4, 18.1, \text{ and } 22.4 \text{ cm}$ from
944	the impact point, (run #151105–3).
945	(b) The relationship between the acceleration, a , and the time in ms, calculated by using
946	Eqs. (14) and (15) under the same impact conditions as those in panel (a). The
947	numbers in the legend represent the $x/R_{\rm rim}$.
948	(c) The relationship between the vibration velocity, V , and the time in ms, calculated
949	with Eqs. (17) and (20) under the same impact conditions as those in panel (a). The
950	numbers in the legend represent the $x/R_{\rm rim}$.
951	(d) The relationship between the vibration velocity, V , and the distance from the impact
952	point normalized by the crater rim radius, x/R_{rim} , calculated with Eqs. (17) and (20)

953 under the same impact conditions as those in panel (a). The numbers in the legend954 represent the elapsed times.

955

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956 Figure 12
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957 (a) The relationship between the seismic energy normalized by the kinetic energy of the

958 projectile, E_s/E_k , and the distance from the impact point normalized by the crater

959 rim radius at the maximum vibration velocity, x_{peak}/R_{rim} , on Fig. 11d under various

960 impact conditions. The open symbols represent the data of loose sand obtained by

961 McGarr et al. [1969].

962 (b) The relationship between the maximum acceleration, g_{max} , and the distance from the

- 963 impact point normalized by the crater rim radius, x/R_{rim} . This figure is same as Fig.
- 8e. The solid circles represent the data of loose sand from McGarr et al. [1969].

965

966	Figure	13
/00		

967 (a) Cross-section of the triaxial accelerometer setting on the target surface. The X-, Y-,

and Z-vectors indicate the directions of the measured acceleration.

969 (b) Impact-induced seismic waves at the three orthogonal directions measured by the

- 970 triaxial accelerometer set at 11 cm from the impact point for a PC projectile at the
- 971 impact velocity of 2.4 km s⁻¹: the crater rim radius is 7.5 cm.
- 972

973 **Figure 14**

974 Comparison of the g_0 values calculated using the semi-theoretical equation of Eq. (27) 975 for various impact conditions (symbols) with those obtained in our experiments (gray 976 zone).

977

978 Figure 15

The relationship between the maximum acceleration, g_{max} , and the distance from the impact point normalized by the crater rim radius, x/R_{rim} , for an impactor with a density of 1.4 g cm⁻³ at the impact velocity of 5 km s⁻¹ on an idealized small (500 m in diameter) body and a 1.46 g cm⁻³ density. The thick lines represent the results calculated with Eqs. (8) and (28) with impactor diameters from 0.01 to 1 m. The horizontal dotted line shows the surface gravity of the idealized small body. The number in each parenthesis represents the crater rim radius.

Run number	Projectile	$v_{\mathrm{i}}, \mathrm{m}\mathrm{s}^{-1}$	D _{rim} , cm	D _c , cm	$d_{\rm rim}$, cm	$d_{\rm c}$, cm	$D_{\rm rim}/D_{\rm c}$	$d_{\rm c}/D_{\rm c}$
150819-3	PC	4230	17.7	13.1	3.8	3.0	1.36	0.23
151104-1	PC	2208	13.7	10.7	2.6	2.1	1.29	0.20
151104-2	PC	6150	20.2	16.0	4.1	3.5	1.26	0.22
151104-3	PC	1125	11.2	8.6	2.3	2.0	1.30	0.23
151105-2	PC	1206	11.7	9.2	2.0	1.6	1.27	0.17
150610-1	Al	1994	8.2	6.3	1.4	1.1	1.30	0.17
150611-3	Al	5015	12.1	9.5	2.8	2.4	1.28	0.26
150612-1	Al	1815	8.2	6.3	1.3	0.9	1.31	0.15
150612-2	Al	2062	8.6	7.0	1.2	0.9	1.24	0.13
150820-1	Al	2920	9.5	7.2	1.5	1.2	1.32	0.16
150820-7	Al	3953	10.8	8.4	1.5	1.2	1.29	0.14
151105-3	Al	4143	10.6	8.2	1.8	1.3	1.29	0.16
150611-5	Gl	5318	11.5	9.3	1.8	1.4	1.24	0.15
150611-6	Gl	2281	8.6	6.9	1.4	1.1	1.25	0.16
150612-5	Gl	2023	8.4	7.0	1.3	1.1	1.19	0.16
150820-2	Gl	4023	10.5	8.4	1.9	1.6	1.24	0.19
150821-1	Gl	3427	10.2	7.9	1.7	1.3	1.28	0.16
150821-2	Gl	4013	10.5	8.6	1.5	1.2	1.23	0.14
150820-3	Ti	3834	11.9	9.8	2.1	1.7	1.21	0.17
150821-3	Ti	2044	10.3	8.2	1.9	1.6	1.25	0.20

986Table 1. Experimental conditions and results on the morphology of impact crater

150821-5	Ti	1972	10.0	7.7	1.9	1.6	1.29	0.20
160720-1	Ti	5181	13.4	11.0	2.1	1.8	1.22	0.16
150820-4	ZrO ₂	3934	13.3	11.4	2.8	2.5	1.17	0.22
160719-3	ZrO ₂	2153	10.8	8.5	2.0	1.6	1.27	0.19
160720-2	ZrO ₂	5727	14.7	11.3	2.9	2.4	1.28	0.21
150610-2	SUS	1841	11.3	8.9	2.6	2.2	1.26	0.25
150611-4	SUS	4936	15.8	12.7	4.1	3.7	1.24	0.29
150612-3	SUS	1848	11.0	8.7	2.7	2.3	1.27	0.27
150612-4	SUS	1824	10.9	8.6	2.7	2.3	1.26	0.27
150820-6	SUS	3805	14.2	11.9	3.1	2.7	1.20	0.23
150820-5	Cu	3626	14.4	11.7	3.7	3.2	1.23	0.27
151105-1	Cu	2114	11.9	9.8	3.1	2.8	1.22	0.29
160719-1	Cu	2238	12.4	9.8	2.9	2.4	1.27	0.25
160720-4	Cu	5348	15.8	12.3	2.7	2.2	1.29	0.18
160721-3	Cu	2306	12.5	10.3	2.6	2.2	1.22	0.22
150819-4	WC	3422	16.6	13.6	4.1	3.6	1.22	0.26

 $D_{\rm rim}$: Crater rim diameter

 D_c : Crater diameter measured on the pre-shot surface

 $d_{\rm rim}$: Crater depth from the rim peak to the crater bottom

 d_c : Crater depth from the pre-shot surface to the crater bottom

 $D_{\rm rim}/D_{\rm c}$: Ratio of crater rim diameter to crater diameter

 d_c/D_c : Ratio of crater depth to crater diameter

Run number	Projectile	$m_{ m p}$, mg	$v_{\rm i},{ m m~s^{-1}}$	R _{rim} , cm	π_2	$\pi_{ m R}$
140528-1	РС	68.0	2038	7.1	5.60×10 ⁻⁹	15.65
140528-2	РС	68.0	2289	7.6	4.44×10 ⁻⁹	16.86
140528-3	РС	68.0	2500	7.5	3.72×10 ⁻⁹	16.63
140528-4	РС	68.0	3852	8.3	1.57×10 ⁻⁹	18.20
140529-1	РС	68.0	1424	6.5	1.15×10 ⁻⁸	14.37
140529-2	РС	68.0	1552	6.7	9.66×10 ⁻⁹	14.70
140529-3	РС	68.0	1684	6.6	8.21×10 ⁻⁹	14.65
140529-4	PC	68.0	1579	6.4	9.34×10 ⁻⁹	14.16
140530-1	PC	68.0	2279	7.2	4.48×10 ⁻⁹	15.88
140530-2	РС	68.0	3748	8.3	1.66×10 ⁻⁹	18.19
140530-3	РС	68.0	5155	9.6	8.76×10 ⁻¹⁰	21.08
140530-4	РС	68.0	3381	8.5	2.04×10 ⁻⁹	18.82
140710-2	PC	68.0	221	3.0	4.76×10 ⁻⁷	6.54
140710-6	РС	68.0	170	2.7	8.04×10 ⁻⁷	5.97
140715-2	РС	68.0	222	3.2	4.71×10 ⁻⁷	6.99
140715-3	РС	68.0	217	3.2	4.92×10 ⁻⁷	7.00
140718-1	PC	68.0	227	3.2	4.50×10 ⁻⁷	7.15
140718-2	PC	68.0	208	3.1	5.36×10 ⁻⁷	6.89
140718-3	PC	68.0	204	3.2	5.59×10 ⁻⁷	7.08
140718-4	РС	68.0	217	3.3	4.92×10 ⁻⁷	7.31

793 Table 2. Experimental conditions, crater size, and non-dimensional scaling parameters

140720-1	PC	68.0	173	2.9	7.78×10 ⁻⁷	6.45
140720-2	PC	68.0	211	3.2	5.23×10 ⁻⁷	7.07
140720-3	PC	68.0	217	3.3	4.97×10 ⁻⁷	7.24
140720-4	PC	68.0	189	3.0	6.49×10 ⁻⁷	6.58
140720-5	PC	68.0	205	3.1	5.52×10 ⁻⁷	6.75
141001-1	PC	68.0	1739	6.6	7.70×10 ⁻⁹	14.65
141002-1	PC	68.0	1506	6.5	1.03×10 ⁻⁸	14.32
141002-2	PC	68.0	2495	7.6	3.74×10 ⁻⁹	16.80
141002-3	PC	68.0	3268	8.7	2.18×10 ⁻⁹	19.08
141002-4	PC	68.0	4417	9.3	1.19×10 ⁻⁹	20.51
141002-5	PC	68.0	5754	10.3	7.03×10 ⁻¹⁰	22.76
141003-1	PC	68.0	6711	11.1	5.17×10 ⁻¹⁰	24.52
141003-2	PC	68.0	6887	11.4	4.91×10 ⁻¹⁰	25.19
141003-3	PC	68.0	4003	9.5	1.45×10 ⁻⁹	21.00
141003-4	PC	68.0	1689	6.8	8.16×10 ⁻⁹	14.97
141003-5	PC	68.0	3016	8.4	2.56×10 ⁻⁹	18.62
141202-1	PC	68.0	6158	10.8	6.14×10^{-10}	23.75
141202-2	PC	68.0	6596	10.8	5.35×10 ⁻¹⁰	23.73
141202-3	PC	68.0	6402	10.9	5.68×10 ⁻¹⁰	24.08
141203-1	PC	68.0	5192	10.3	8.63×10 ⁻¹⁰	22.60
141203-2	PC	68.0	5330	9.9	8.19×10 ⁻¹⁰	21.85
141203-3	PC	68.0	2385	7.5	4.09×10 ⁻⁹	16.62
141203-4	PC	68.0	2254	7.6	4.58×10 ⁻⁹	16.66

141204-1	PC	68.0	2317	7.3	4.34×10 ⁻⁹	16.16
150819-1	PC	68.0	4322	9.7	1.25×10 ⁻⁹	21.33
*150819-3	РС	68.0	4230	8.9	1.30×10 ⁻⁹	19.52
*151104-1	PC	68.0	2208	6.9	4.77×10 ⁻⁹	15.17
*151104-2	PC	68.0	6150	10.1	6.15×10 ⁻¹⁰	22.29
*151104-3	PC	68.0	1125	5.6	1.84×10^{-8}	12.32
151105-2	PC	68.0	1206	5.8	1.60×10 ⁻⁸	12.87
#150309-1	Al	11.6	4124	5.6	5.76×10 ⁻¹⁰	22.14
#150310-1	Al	11.6	2077	4.4	2.27×10 ⁻⁹	17.62
#150310-2	Al	11.6	2059	4.5	2.31×10 ⁻⁹	17.67
#150610-1	Al	11.6	1994	4.2	2.46×10 ⁻⁹	16.52
150611-2	Al	11.7	4936	6.1	4.02×10^{-10}	24.14
150611-3	Al	11.7	5015	6.3	3.90×10 ⁻¹⁰	24.88
#150612-1	Al	11.9	1815	4.0	2.97×10 ⁻⁹	15.70
#150612-2	Al	11.5	2062	4.4	2.30×10 ⁻⁹	17.60
150820-1	Al	11.7	2920	4.7	1.15×10 ⁻⁹	18.77
#150820-7	Al	11.8	3953	5.4	6.27×10 ⁻¹⁰	21.43
* ^{, #} 151105-3	Al	11.7	4143	5.3	5.71×10 ⁻¹⁰	20.94
160721-2	Al	11.9	5622	6.1	3.10×10 ⁻¹⁰	24.11
150611-5	Gl	10.6	5318	5.8	3.47×10 ⁻¹⁰	23.93
150611-6	Gl	10.6	2281	4.4	1.88×10 ⁻⁹	18.19
#150612-5	Gl	11.3	2023	4.2	2.39×10 ⁻⁹	16.86
* ^{,#} 150820-2	Gl	10.9	4023	5.2	6.06×10 ⁻¹⁰	21.24

150821-1	Gl	10.7	3427	5.1	8.34×10 ⁻¹⁰	20.77
#150821-2	Gl	10.6	4013	5.3	6.09×10 ⁻¹⁰	21.58
#150311-1	Ti	18.9	4097	6.5	5.84×10 ⁻¹⁰	22.00
* ^{, #} 150820-3	Ti	18.9	3834	5.9	6.67×10 ⁻¹⁰	20.06
#150821-3	Ti	18.8	2044	5.1	2.35×10 ⁻⁹	17.37
#150821-4	Ti	18.8	1972	5.0	2.52×10 ⁻⁹	16.84
160720-1	Ti	18.8	5181	6.7	3.65×10 ⁻¹⁰	22.68
#150309-3	ZrO ₂	26.5	3864	6.9	6.56×10 ⁻¹⁰	20.79
#150311-7	ZrO ₂	25.7	2004	5.7	2.44×10 ⁻⁹	17.31
*,#150820-4	ZrO ₂	25.4	3934	6.7	6.33×10 ⁻¹⁰	20.42
#160719-3	ZrO ₂	26.9	2153	5.4	2.11×10 ⁻⁹	16.26
160720-2	ZrO ₂	24.2	5727	7.3	2.99×10 ⁻¹⁰	22.85
#150309-2	SUS	32.4	3943	7.5	6.30×10 ⁻¹⁰	21.14
#150310-3	SUS	32.6	2018	5.7	2.41×10 ⁻⁹	16.01
#150610-2	SUS	32.6	1841	5.7	2.89×10 ⁻⁹	16.12
150611-4	SUS	32.6	4936	8.4	4.02×10^{-10}	23.55
#150612-3	SUS	32.6	1848	5.7	2.87×10 ⁻⁹	15.98
#150612-4	SUS	32.8	1824	5.8	2.95×10 ⁻⁹	16.18
* ^{, #} 150820-6	SUS	32.6	3805	7.1	6.77×10 ⁻¹⁰	20.07
#150310-6	Cu	37.8	3962	7.6	6.24×10 ⁻¹⁰	20.32
*150820-5	Cu	38.1	3626	7.2	7.45×10 ⁻¹⁰	19.33
#151105-1	Cu	37.8	2114	6.0	2.19×10 ⁻⁹	16.01
160719-1	Cu	38.2	2238	6.2	1.96×10 ⁻⁹	16.60

160720-4	Cu	37.9	5348	7.9	3.43×10 ⁻¹⁰	21.10
160721-3	Cu	37.9	2306	6.3	1.84×10 ⁻⁹	16.75
150309-4	WC	62.1	3726	9.1	7.06×10 ⁻¹⁰	20.63
150312-2	WC	61.9	1786	6.8	3.07×10 ⁻⁹	15.47
*150819-4	WC	62.5	3422	8.3	8.37×10^{-10}	18.82

994 The asterisk (*) indicates the example of the crater profile shown on Fig. 3.

995 The sharp (#) indicates the data calculating by Eq. (5) shown on Fig. 5.

- $m_{\rm p}$: projectile mass
- v_i : impact velocity
- $R_{\rm rim}$: crater rim radius
- π_2 : a scaling parameter defined by Eq. (2)
- $\pi_{\rm R}$: a scaling parameter defined by Eq. (1)

Run number	$v_{\rm i}$, m s ⁻¹	E _k , J	R _{rim} , cm	x, cm	x/R _{rim}	$t_{ m max},$ ms	$g_{\rm max}$, m s ⁻²	T _{half} , ms
140710-2	221	1.7	3.0	5.1	1.6	0.42	30.0	1.13
				7.5	2.5	0.87	7.1	0.63
140715-2	222	1.67	3.2	6.0	1.8	0.39	35.5	0.79
140718-2	208	1.5	3.1	5.7	1.7	0.87	18.1	1.52
				6.6	2.0	0.70	12.4	1.02
				7.2	2.2	1.01	6.75	1.66
140718-3	204	1.4	3.2	8.8	2.7	1.32	7.13	1.59
140718-4	217	1.6	3.3	9.4	2.8	1.31	9.6	0.78
140720-1	173	1.0	2.9	6.4	2.1	0.66	10.5	1.02

1001 Table 3. Experimental conditions and results of a PC projectile at low impact velocity.

 v_i : Impact velocity

 E_k : Kinetic energy of the projectile

 $R_{\rm rim}$: Crater rim radius

x: Distance from the impact point to the accelerometer

 t_{max} : Traveling time from the impact time to the time when the acceleration becomes the maximum

 g_{max} : Maximum acceleration

 T_{half} , Duration of the first upward acceleration

Run number	v_{i} , m s ⁻¹	E _k , J	R _{rim} , cm	x, cm	$x/R_{\rm rim}$	$t_{ m max}$, ms	$g_{\rm max}$, m s ⁻²	T _{half} , ms
140530-1	2279	176.6	7.6	10.3	1.4	0.93	67.3	1.25
				16.2	2.1	1.74	17.0	1.44
				25.1	3.3	3.07	2.8	1.66
140530-2	3748	477.6	8.7	12.3	1.4	0.94	116.4	1.02
				17.2	2.0	1.83	27.6	1.90
				25.1	2.9	2.81	8.2	1.50
140530-3	5155	903.5	10.2	15.2	1.5	1.26	116.2	1.24
				20.2	2.0	2.03	22.0	1.28
				27.1	2.7	3.49	15.3	2.33
140530-4	3381	388.7	8.9	15.2	1.7	1.64	57.2	0.89
				20.2	2.3	2.41	21.9	1.55
				30.1	3.4	3.59	9.7	1.32
141001-1	1739	102.8	6.6	10.7	1.6	0.97	45.6	1.77
				14.0	2.1	1.56	25.1	1.64
141002-1	1506	77.1	6.5	8.9	1.4	—	35.0	1.17
				12.0	1.9	—	30.7	1.81
				15.2	2.3	—	9.7	1.52
141002-2	2495	211.7	7.6	12.1	1.6	1.05	56.9	2.48
				14.8	1.9	1.72	30.1	2.46
141002-3	3268	363.1	8.6	18.9	2.2	2.30	28.4	2.04

1009 Table 4. Experimental conditions and results of a PC projectile at high impact velocity.

141002-4	4417	663.3	9.3	16.6	1.8	1.77	44.0	2.53
				20.6	2.2	2.56	13.9	2.57
141002-5	5754	1125.7	10.3	15.4	1.5	1.58	69.5	1.94
				20.2	2.0	2.38	13.8	1.68
141003-1	6711	1531.3	11.1	15.0	1.3	1.27	77.7	2.46
				20.6	1.9	2.16	34.3	2.66
141003-2	6887	1612.7	11.4	13.3	1.2	0.97	118.2	1.23
				17.2	1.5	1.59	79.7	2.05
				23.8	2.1	3.10	21.6	2.24
141003-3	4003	544.8	9.5	13.6	1.4	1.25	63.7	1.87
				17.6	1.8	2.10	25.6	1.88
				23.6	2.5	3.09	5.6	2.28
141003-4	1689	97.0	6.8	7.9	1.2	0.42	46.4	1.78
				15.3	2.2	1.99	28.9	2.10
				23.1	3.4	3.07	5.6	2.18
141003-5	3016	309.3	8.4	10.1	1.2	0.81	53.0	1.75
				15.7	1.9	2.12	28.8	2.09
				19.9	2.4	2.98	11.4	2.48
141202-1	6158	1289.3	10.8	20.1	1.9	1.97	30.0	1.94
				23.1	2.1	2.42	14.2	2.15
141202-2	6596	1479.3	10.8	15.3	1.4	1.67	47.8	2.61
				17.6	1.6	1.97	34.5	2.60

-	÷.		÷					
				24.6	2.3	3.59	14.2	2.40
141202-3	6402	1393.5	10.9	14.6	1.3	1.85	35.9	2.27
				19.8	1.8	2.46	33.5	2.16
				24.2	2.2	3.78	15.7	2.15
141203-1	5192	916.5	10.2	14.3	1.4	1.84	53.5	1.99
				17.1	1.7	2.12	27.9	2.04
				23.4	2.3	3.51	16.1	1.55
141203-2	5330	965.9	9.9	12.1	1.2	1.13	77.8	2.17
				16.7	1.7	2.20	26.0	2.52
				19.6	2.0	2.91	20.9	2.29
				19.9	2.0	2.91	10.5	2.60
150819-1	4322	635.1	9.7	11.6	1.2	0.88	86.6	1.91
				17.8	1.8	2.03	27.6	1.95
				23.6	2.4	2.88	14.8	1.78
				26.0	2.7	3.38	9.1	1.61
150819-3	4230	608.4	8.9	12.9	1.5	1.27	68.0	2.13
				15.7	1.8	1.86	36.0	2.28
				20.6	2.3	2.62	19.7	2.44
				26.2	3.0	3.75	11.2	2.21
151104-1	2208	165.8	6.9	12.4	1.8	1.51	47.3	1.48
				13.9	2.0	1.89	27.3	1.89
				15.2	2.2	2.24	19.3	1.52

				19.2	2.8	2.74	9.7	1.58
				23.3	3.4	3.27	3.7	2.14
151104-3	1125	43.0	5.6	12.1	2.2	1.53	7.9	1.66
				15.6	2.8	2.13	3.6	2.66
				19.2	3.4	2.78	2.8	1.97
151105-2	1206	49.5	5.8	10.3	1.8	0.96	6.9	2.94
				13.0	2.2	1.48	6.5	2.05
				16.5	2.8	2.13	5.3	1.72
				18.3	3.1	2.52	5.2	1.74
				21.5	3.7	3.06	4.0	1.54

Run number	Projectile	$v_{i}, m s^{-1}$	$E_{\rm k}, { m J}$	R _{rim} , cm	x, cm	$x/R_{\rm rim}$	t _{max} , ms	$g_{\rm max}$, m s ⁻²	T _{half} , ms
150611-5	Gl	5318	150.3	5.8	11.1	1.9	1.45	14.7	2.39
					15.1	2.6	2.29	5.6	2.19
					21.0	3.6	3.42	1.5	2.07
					23.0	3.9	3.36	1.6	1.82
150611-6	Gl	2281	28.0	4.4	13.1	3.0	2.16	3.1	2.12
					18.2	4.1	2.37	0.5	1.49
					23.3	5.2	3.59	0.5	1.77
					23.9	5.4	4.17	0.4	1.58
150612-5	Gl	2023	23.1	4.2	9.2	2.2	1.45	8.9	2.13
					17.3	4.1	2.39	0.5	1.11
150820-2	Gl	4023	88.2	5.0	14.1	2.8	2.07	7.1	2.10
					17.9	3.6	2.77	5.0	1.94
					21.9	4.4	3.43	2.4	1.85
150821-1	Gl	3427	62.5	5.0	14.2	2.8	1.96	5.0	1.96
					16.2	3.2	2.49	5.1	1.93
					23.2	4.6	3.58	0.9	1.87
150821-2	Gl	4013	85.5	5.1	13.2	2.6	1.93	7.4	2.81
					16.7	3.3	2.66	4.0	2.29
					20.4	4.0	3.33	2.3	2.15
150309-1	Al	4124	98.8	5.7	9.6	1.7	0.99	36.2	1.82

1011 Table 5. Experimental conditions and results of 2-mm projectiles with different densities.

					12.7	2.2	1.97	12.4	1.40
					12.8	2.2	1.49	11.1	1.18
					14.8	2.6	2.01	8.0	1.13
					19.3	3.4	2.48	5.4	1.29
150310-2	Al	2059	24.7	4.4	10.6	2.4	1.86	7.9	1.40
					21.9	4.9	3.85	0.9	1.60
150610-1	Al	1994	23.1	4.2	7.9	1.9	0.86	12.7	0.99
					9.9	2.4	1.13	11.3	1.07
					13.9	3.3	1.87	3.8	1.36
					17.3	4.2	2.16	2.5	1.45
					24.2	5.8	3.61	1.4	1.42
150611-2	Al	4936	142.3	6.1	13.2	2.2	1.78	7.5	1.88
					17.7	2.9	2.49	4.1	1.74
					20.1	3.3	2.65	2.1	1.81
					25.6	4.2	3.48	1.1	1.64
150611-3	Al	5015	146.9	6.3	13.5	2.2	1.70	8.2	1.72
					18.3	2.9	2.53	4.2	1.25
					20.5	3.3	2.72	3.1	1.54
150612-1	Al	1815	19.6	4.0	16.0	4.0	2.25	1.5	1.40
					22.2	5.6	3.00	0.7	1.33
150612-2	Al	2062	24.5	4.4	21.9	5.0	2.69	0.3	1.52
150820-1	Al	2920	49.9	4.6	12.0	2.6	1.33	14.5	1.83
					15.3	3.3	1.92	5.8	1.66

					18.5	4.0	2.68	5.6	1.79
150820-7	Al	3953	92.0	5.2	12.2	2.3	1.76	8.0	1.88
					17.5	3.3	2.68	2.7	2.14
151105-3	Al	4143	100.3	5.4	10.7	2.0	1.30	13.6	2.34
					13.0	2.4	2.01	6.2	1.72
					15.4	2.8	2.46	3.5	1.68
					18.1	3.3	2.82	3.8	1.73
					22.4	4.1	3.33	2.6	1.53
150311-1	Ti	4097	159.0	6.5	13.4	2.1	1.88	10.7	2.38
					15.9	2.4	2.32	5.1	2.51
					18.1	2.8	2.84	6.5	2.08
					20.5	3.1	3.50	1.8	1.66
150820-3	Ti	3834	138.8	5.8	12.1	2.1	_	16.5	2.53
					18.4	3.2	-	9.8	2.06
					24.6	4.2	-	3.5	1.73
150821-5	Ti	1972	36.6	4.8	14.0	2.9	2.04	4.0	1.48
					15.1	3.2	2.39	4.1	1.92
					19.8	4.2	3.22	1.8	1.89
150309-3	ZrO ₂	3864	197.8	6.9	10.0	1.4	0.94	24.0	2.43
					15.3	2.2	2.21	15.8	2.31
					18.2	2.6	2.67	4.4	2.03
					19.1	2.8	2.82	6.2	1.82
					22.6	3.3	3.61	3.2	1.67

r			1	1	1			1	1
150311-7	ZrO ₂	2004	51.5	5.7	12.5	2.2	1.83	12.7	1.80
150820-4	ZrO ₂	3934	196.5	6.6	11.7	1.8	1.23	15.4	2.76
					15.0	2.3	1.94	7.5	2.53
					17.2	2.6	2.51	12.2	2.09
					22.1	3.3	3.21	5.3	1.77
150309-2	SUS	3943	252.1	7.5	14.9	2.0	2.36	12.9	2.28
					16.4	2.2	2.73	8.7	2.43
					17.7	2.4	2.67	8.8	2.06
					21.1	2.8	3.56	3.5	1.98
150310-3	SUS	2018	66.4	5.7	17.4	3.1	3.19	3.3	2.10
150610-2	SUS	1841	55.2	5.7	11.6	2.0	1.78	11.7	1.87
					15.9	2.8	2.53	3.3	2.05
					21.2	3.7	3.10	1.2	1.60
					25.3	4.4	3.80	1.1	1.71
150611-4	SUS	4936	397.5	8.4	15.6	1.9	1.85	14.9	2.21
					18.1	2.2	2.77	7.7	2.09
					21.6	2.6	3.32	4.6	2.52
					24.8	3.0	3.92	7.6	2.28
150612-3	SUS	1848	55.6	5.7	15.6	2.7	2.25	6.6	1.94
					21.6	3.8	3.27	2.0	1.82
150612-4	SUS	1824	54.5	5.8	11.8	2.1	1.64	9.1	1.95
150820-6	SUS	3805	235.6	7.0	12.7	1.8	1.44	26.0	2.71
					17.3	2.5	2.49	22.1	2.17

					21.0	3.0	3.25	8.2	1.82
					24.0	3.4	3.51	6.5	1.55
150310-6	Cu	3962	296.4	7.6	14.8	2.0	1.98	15.4	2.00
					17.1	2.3	2.55	14.1	2.36
					20.0	2.6	2.88	7.7	2.00
					23.4	3.1	3.61	6.0	1.50
150820-5	Cu	3626	250.1	7.1	13.9	2.0	1.45	20.0	2.20
					18.4	2.6	2.76	14.2	1.96
					25.0	3.5	3.38	7.5	1.67
151105-1	Cu	2114	84.5	6.0	9.9	1.6	0.87	28.5	1.09
					14.1	2.3	1.54	10.0	1.49
					17.2	2.9	2.31	5.8	1.15
					19.7	3.3	2.75	7.1	1.20
					24.8	4.1	3.41	3.1	1.44
150309-4	WC	3726	430.7	9.1	14.0	1.5	1.42	40.9	2.06
					15.8	1.7	1.87	37.0	1.99
					20.6	2.3	2.79	13.0	1.99
					21.2	2.3	2.81	12.9	2.02
					24.4	2.7	3.76	8.4	2.19
150312-2	WC	1786	98.8	6.4	12.2	1.9	2.13	12.4	2.70
					15.8	2.5	3.39	8.5	2.80
150819-4	WC	3422	366.2	8.2	12.7	1.6	1.16	46.3	2.89
					15.6	1.9	1.68	16.5	2.95

		18.5	2.3	2.46	20.8	2.65
		26.2	3.2	3.75	6.8	2.05
Impact	Propagation	Average crater	Eq. (7) related to maximum			
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velocity v_i ,	velocity V_{prop} ,	rim radius	acceleration g_{max}			
$\rm km~s^{-1}$	$m s^{-1}$	$R_{\rm rim_ave},{\rm cm}$	a _r	$b_{ m r}$		
0.2	37.4 (7.3)	3.1	-1.68 (0.88)	2.38 (0.76)		
1	53.4 (2.9)	5.7	-0.18 (0.30)	1.08 (0.36)		
1.5	56.4 (3.9)	6.6	-0.35 (0.41)	1.92 (0.46)		
2	57.2 (7.1)	7.6	-1.77 (0.16)	3.78 (0.20)		
3	65.0 (10.2)	8.9	0.09 (0.37)	1.75 (0.48)		
4	56.5 (3.0)	9.0	-0.92 (0.19)	3.14 (0.25)		
5	47.9 (8.1)	10.1	-0.46 (0.49)	2.59 (0.66)		
6	40.7 (12.9)	10.7	-0.39 (0.60)	2.52 (0.84)		
7	45.3 (4.5)	11.1	-0.54 (0.38)	2.93 (0.51)		

1016 The number in the parentheses indicates the standard error.

	Impact Propagation		Average crater	Eq. (7) related to maximum	
Projectile	velocity v_i ,	velocity V_{prop} ,	rim radius	acceleration g_{max}	
	$km s^{-1}$	$m s^{-1}$	$R_{\rm rim_ave},{\rm cm}$	a _r	$b_{ m r}$
Al	2	56.1 (7.5)	4.2	-2.08 (0.39)	3.00 (0.48)
	3	47.5 (4.0)	4.6	-0.98 (0.86)	2.27 (1.03)
	4	53.4 (5.6)	5.5	-1.42 (0.38)	2.70 (0.45)
	5	70.6 (3.9)	6.2	-1.56 (0.25)	2.84 (0.33)
Gl	2	54.0 (9.1)	4.3	-2.64 (0.45)	3.41 (0.57)
	3	56.6 (6.7)	5.0	-2.33 (0.77)	3.68 (1.01)
	4	55.0 (3.0)	5.1	-1.21 (0.32)	2.40 (0.41)
	5	56.8 (7.6)	5.8	-1.95 (0.25)	3.26 (0.31)
Ti	2	50.3 (6.0)	4.8	-1.43 (0.43)	2.43 (0.54)
	4	43.0 (3.3)	7.3	-1.01 (0.61)	2.35 (0.78)
ZrO ₂	4	48.9 (2.4)	6.8	-0.76 (0.35)	2.15 (0.44)
SUS	2	60.4 (7.7)	5.7	-1.81 (0.28)	3.04 (0.36)
	4	49.8 (7.3)	7.3	-0.81 (0.62)	2.41 (0.83)
	5	45.2 (6.0)	8.4	-0.27 (0.84)	1.66 (1.19)
Cu	2	56.4 (3.8)	6.0	-0.87 (0.26)	2.27 (0.33)
	4	52.4 (7.7)	7.3	-0.34 (0.26)	1.90 (0.35)
WC	3	51.5 (3.1)	8.2	-0.54 (0.51)	2.39 (0.66)
	4	46.1 (2.9)	9.1	-0.96 (0.20)	3.06 (0.27)

1017 Table 7. Propagation velocity, parameters related to maximum acceleration on Eq. (7) for 2-mm projectiles with different densities.

1018 The number in the parentheses indic1ates the standard error.















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Figure 10

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Figure 13







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Figure 15